The Geometry of Classical and Quantum Fields

Markus J. Pflaum

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Authors

The following people have contributed to this work, in alphabetical order:

JONATHAN BELCHER MARKUS J. PFLAUM (main author) DANIEL SPIEGEL

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Attribution

The main author of this work is MARKUS J. PFLAUM.

 $\rm JONATHAN$ $\rm BELCHER$ contributed notes of lectures held by $\rm M.J.$ $\rm PFLAUM.$ From these notes the following material has been incorporated:

- Remark 1.1.8,
- in Section 1.2, the statement of Theorem 1.2.5 (Wigner's theorem), and Theorem 1.2.8 (Bargmann's theorem),
- Proposition 6.3.3.

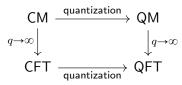
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Sections ??, Orthonormal bases in Hilbert spaces, and A.3.5, Adjoint operators, are based on notes taken by DANIEL SPIEGEL of a lecture series by M.J. PFLAUM.

Introduction

Classical and quantum mechanical systems are mathematically described in a different way. For finitely many degrees of freedom, differential geometry, notably symplectic and Poisson geometry, provides the language in which classical mechanical systems are described, whereas functional analysis and in particular the theory of Hilbert spaces is the appropriate language in which quantum mechanics is formulated. The mathematics is well understood in both situations, and one even has a powerful tool for the passage from the classical to the quantum mechanical description of a corresponding system, namely quantization theory.

In their book (*Mathematical Concepts of Quantum Mechanics*, Gustafson & Sigal, 2011), the authors depict the situation by the following diagram, where $d \rightarrow \infty$ denotes the passage from finitely to infinitely many degrees of freedom.



The key ingrediants for the description of a physical system are the mathematical objects which encode its state space, the observable space, and its dynamics. These objects should depend in some functorial on the system and usually come from quite distinct categories, depending on whether the system is classical or quantum, has finitely or infinitely many degrees of freedom.

Part I. Classical Field Theory

I.1. Variational calculus

1.1. The variational bicomplex

The Cartan distribution

1.1.1 We start with a smooth fiber bundle $\pi : E \to M$ over a *d*-dimensional manifold M. The typical fiber is denoted F and assumed to have dimension n. Consider the *infinite jet bundle* $\pi_{\infty} : J^{\infty}E \to M$ and recall that $(J^{\infty}E, \mathbb{C}^{\infty})$ is the pro-manifold defined as the limit of the (cofiltered) diagram

$$(E, \mathbb{C}^{\infty}) \stackrel{\pi_{0,1}}{\leftarrow} (\mathsf{J}^{1}E, \mathbb{C}^{\infty}) \stackrel{\pi_{1,2}}{\leftarrow} \dots \stackrel{\pi_{k-1,k}}{\leftarrow} (\mathsf{J}^{k}E, \mathbb{C}^{\infty}) \stackrel{\pi_{k,k+1}}{\leftarrow} \dots$$
 (1.1.1)

in the category of commutative locally \mathbb{R} -ringed spaces. This means that in the category of topological spaces $J^{\infty}E$ coincides with $\lim_{k\in\mathbb{N}} J^kE$ and that the structure sheaf $\mathcal{C}_{J^{\infty}E}^{\infty}$ is given by $\underset{k\in\mathbb{N}}{\operatorname{colim}} \pi_{k,\infty}^* \mathcal{C}_{J^kE}^{\infty}$, where the $\pi_{k,\infty}: J^{\infty}E \to J^kE$ are the natural maps from the (topological) limit to the objects of the diagram. The projection $\pi_{\infty}: J^{\infty}E \to M$ is uniquely determined by the property that $\pi_{\infty} = \pi_k \circ \pi_{k,\infty}$ for all $k \in \mathbb{N}$, where π_k is the canonical projections of the finite jet bundles J^kE . Note that the family of canonical projections $\pi_k: J^kE \to M$ is compatible with the diagram Equation (1.1.1) in the sense that $\pi_l = \pi_k \circ \pi_{k,l}$ for all $k \leq l$. Next recall that $\mathcal{C}_{\mathrm{loc},J^{\infty}E}^{\infty}$, or just $\mathcal{C}_{\mathrm{loc}}^{\infty}$ when no confusion can arise, stands for the presheaf of *local functions* on the infinite jet bundle. Its space of sections over some open $U \subset J^{\infty}E$ consists of all continuous maps $f: U \to \mathbb{R}$ for which there exists a $k \in \mathbb{N}$, an open $U_k \subset J^kE$ and a smooth function $f_k: U_k \to \mathbb{R}$ such that $U \subset \pi_{k,\infty}^{-1}(U_k)$ and $f = f_k \circ \pi_{k,\infty}|_U$.

The diagram Equation (1.1.1) of jet bundles of finite order induces another filtered diagram by taking tangent bundles and tangent maps:

$$(TE, \mathbb{C}^{\infty}) \stackrel{T\pi_{0,1}}{\longleftarrow} (T\mathsf{J}^{1}E, \mathbb{C}^{\infty}) \stackrel{T\pi_{1,2}}{\longleftarrow} \dots \stackrel{T\pi_{k-1,k}}{\longleftarrow} (T\mathsf{J}^{k}E, \mathbb{C}^{\infty}) \stackrel{T\pi_{k,k+1}}{\longleftarrow} \dots$$
 (1.1.2)

The resulting limit in the category of commutative locally \mathbb{R} -ringed spaces is called the *tangent bundle* of the pro-manifold $(J^{\infty}E, \mathbb{C}^{\infty})$ and is denoted $(TJ^{\infty}E, \mathbb{C}^{\infty})$. One writes $T\pi_{k,\infty}: TJ^{\infty}E \to TJ^{k}E$ for the natural maps of the limit and obtains the tangent map $T\pi_{\infty}: TJ^{\infty}E \to TM$ uniquely determined by the property that $T\pi_{\infty} = T\pi_{k} \circ T\pi_{k,\infty}$ for all $k \in \mathbb{N}$.

As the last prerequisite we need the concept of Roman multi-indices and their combinatorial properties from Section A.8.1. As there we denote by $\overline{\mathcal{I}}^{\bullet}$ the set of *ordered Roman multi-indices* in an ordered index set \mathcal{I} . In our situation, the index set is $\mathcal{I} = \{1, \ldots, d\}$ which entails that $\overline{\mathcal{I}}^{\bullet}$ consists of a zero element O and all finite sequences of integers of the form

$$I = (i_1, \ldots, i_k)$$
, where $k \in \mathbb{N}_{>0}$ and $1 \leq i_1 \leq i_2 \leq \ldots \leq i_k \leq d$.

The number k is called the *length* of the ordered Roman multi-index I and is denoted by |I|. The *length* of the zero element O is defined to be 0.

Now we have all the tools to define the main object of this section, the Cartan distribution.

1.1.2 Let p be a point of the base manifold M. Choose an open contractible neighborhood $U \subset M$ of p over which there exists a coordinate system $x : U \to \mathbb{R}^d$. Denote by \mathcal{E}_U the space of smooth sections of the bundle $\pi : E \to M$ over U and by I_{ε} for $\varepsilon > 0$ the open interval $(-\varepsilon, \varepsilon)$ around 0. By Borel's theorem, the jet map $j_q^{\infty} : \mathcal{E}_U \to J_q^{\infty} E$ is surjective for every $q \in U$. Call a smooth path

$$\gamma = (\sigma, \mu) : I_{\varepsilon} \to \mathcal{E}_U \times U, \ t \mapsto (\sigma_t, \mu_t)$$

with $\mu_0 = p$ vertical over p if μ is a constant path and horizontal over p if σ is a constant path. Smoothness of σ hereby means that $\sigma^{\vee} : I_{\varepsilon} \times U \to E$, $(t,q) \mapsto \sigma_t(q)$ is smooth. The composition

$$\mathbf{j}^{\infty} \circ \gamma : I_{\varepsilon} \to \mathbf{J}^{\infty} E, \ t \mapsto \mathbf{j}_{\mu_t}^{\infty}(\sigma_t)$$

then is a smooth path in the jet bundle and the derivative

$$\left(\mathbf{j}^{\infty} \circ \gamma\right)'(0) = \frac{d}{dt} \left(\mathbf{j}^{\infty} \circ \gamma\right)(t) \Big|_{t=0} = \frac{d}{dt} \left(\mathbf{j}^{\infty}_{\mu_t}(\sigma_t)\right) \Big|_{t=0}$$

an element of the tangent space $T_{\theta} J^{\infty} E$ over the footpoint $\theta = j_p^{\infty}(\sigma_0)$. If γ is vertical, the path $\pi_{\infty} \circ j^{\infty} \circ \gamma$ is constant with value p which implies that the tangent vector $(j^{\infty} \circ \gamma)'(0)$ has to be an element of the vertical bundle $\nabla \pi_{\infty} = \ker T \pi_{\infty} \subset T J^{\infty} E$. Let us show that every vertical tangent vector with footpoint θ can be obtained that way. So assume that $v \in V_{\theta}\pi_{\infty}$ is represented by a smooth path $\varrho : (-\varepsilon, \varepsilon) \to J^{\infty} E$ such that $\pi_{\infty}(\varrho(t)) = p$ for all t. After possibly shrinking U and ε one can assume that there exists a fibered chart $(x, u) : \tilde{U} \to \mathbb{R}^d \times \mathbb{R}^n$ over some open $\tilde{U} \subset E$ such that $\pi(\tilde{U}) = U$, (x, u) is trivialising in the sense that its image coincides with the cartesian product of x(U) and an open $V \subset \mathbb{R}^n$ and such that $\pi_{0,\infty}(\varrho(t)) \in \tilde{U}$ for all t. One obtains a family of smooth real valued functions $u^a \circ \varrho, u_i^a \circ \varrho, \ldots, u_1^a \circ \varrho, \ldots$, where the index a runs through $\{1, \ldots, n\}$, the index i through $\mathfrak{I} = \{1, \ldots, d\}$, and I through all ordered Roman multi-indices in \mathfrak{I} of order ≥ 2 . By Borel's Theorem with parameters (Kriegl & Michor, 1997, 15.4), there exists a smooth function $s = (s^1, \ldots, s^n) : I_{\varepsilon} \times U \to V$ such that

$$\frac{\partial^{|I|}s^{\mathrm{a}}}{\partial x^{\mathrm{I}}}(t,p) = u^{\mathrm{a}}_{\mathrm{I}} \circ \varrho(t) \quad \text{for all } t \in I_{\varepsilon}, \ \mathrm{a} \in \{1,\ldots,n\} \text{ and } \mathrm{I} \in \overline{\mathfrak{I}}^{\bullet} \ .$$

Let $\sigma: I_{\varepsilon} \to \mathcal{E}_U$ be the smooth path of sections $t \mapsto s(t, -)$, $\mu: I_{\varepsilon} \to M$ the constant path at p and let $\gamma = (\sigma, \mu)$. Then γ is vertical and, by construction,

$$(\mathbf{j}^{\infty} \circ \gamma)'(0) = \varrho'(0) = v .$$

This shows the claim.

Next assume to be given a jet $\theta \in J_p^{\infty} E$. Define the *horizontal space* at that jet by

$$\mathsf{C}_{\theta}\mathsf{J}^{\infty}E = \left\{ \left(\mathsf{j}^{\infty} \circ \gamma\right)'(0) \in T_{\theta}\mathsf{J}^{\infty}E \mid \gamma = (\sigma, \mu) \text{ is horizontal over } p \text{ and } \mathsf{j}^{\infty}\sigma_0 = \theta \right\}.$$

One calls $CJ^{\infty}E = \bigcup_{\theta \in J^{\infty}E} C_{\theta}J^{\infty}E$ the *Cartan distribution* on the jet bundle $J^{\infty}E$. In the following we will study its properties and will show that it is an involutive distribution on the jet bundle which is complementary to the vertical bundle.

1.1.3 Lemma Let $\theta \in J_p^{\infty} E$ be a jet and choose a trivialising fibered chart $(x, u) : \widetilde{U} \to \mathbb{R}^d \times \mathbb{R}^n$ around an open neighborhood of $e = \pi_{1,\infty}(\theta)$. Let $\mu : I_{\varepsilon} \to U$ be a smooth path with $\mu_0 = p$, $\sigma : I_{\varepsilon} \to \mathcal{E}_U$ a smooth path of sections and finally $s : U \to E$ a smooth section such that that the images of all σ_t and s are in \widetilde{U} and such that $j_p^{\infty}(\sigma_0) = j_p^{\infty}(s) = \theta$. Denote by μ^i the composition $x^i \circ \mu$ and by σ^a and s^a the compositions $u^a \circ \sigma$ and $u^a \circ s$, respectively. Then the tangent vector of the vertical path (σ, p) is given by

$$\left(\mathbf{j}_{p}^{\infty}\sigma_{t}\right)'(0) = \sum_{\mathbf{a}=1}^{n}\sum_{\mathbf{I}}\frac{\partial^{|\mathbf{I}|}(\sigma^{\mathbf{a}})'(0)}{\partial x^{\mathbf{I}}}(p)\frac{\partial}{\partial u_{\mathbf{I}}^{\mathbf{a}}}$$
(1.1.3)

and the tangent vector of the horizontal path (s, μ) by

$$\left(\mathbf{j}_{\mu_{t}}^{\infty}s\right)'(0) = \sum_{i=1}^{d} (\mu^{i})'(0) \left(\frac{\partial}{\partial x^{i}} + \sum_{a=1}^{n} \sum_{\mathbf{I}} \frac{\partial^{|I|+1}s^{a}}{\partial x^{i}\partial x^{\mathbf{I}}}(p)\frac{\partial}{\partial u_{\mathbf{I}}^{a}}\right) .$$
(1.1.4)

In these formulas, I runs through all ordered Roman multi-indices in the index set $\mathcal{I} = \{1, \dots, d\}$. *Proof.* Let $\gamma = (\sigma, \mu)$. Then in the selected fibered chart

$$x^{i} \circ j^{\infty} \circ \gamma = \mu^{i}$$
 and $(u_{I}^{a} \circ j^{\infty} \circ \gamma)(t) = \frac{\partial^{|I|} \sigma_{t}^{a}}{\partial x^{i}}(\mu_{t})$

from which the claim follows by specialization to $\mu_t = p$ respectively $\sigma_t = s$ and the chain rule. \Box

1.1.4 Lemma Let $\theta \in J_p^{\infty}E$ be a jet and $s_1, s_2: U \to E$ two smooth sections such that

$$\theta = \mathsf{j}_p^\infty(s_1) = \mathsf{j}_p^\infty(s_2)$$
 .

Then for every smooth path $\mu: I_{\varepsilon} \to M$ with $\mu_0 = p$ the equality

$$\left(\mathsf{j}_{\mu_t}^{\infty}s_1\right)'(0) = \left(\mathsf{j}_{\mu_t}^{\infty}s_2\right)'(0)$$

holds true, where ' denotes the derivative with respect to the parameter t. Hence,

$$C_{\theta} \mathsf{J}^{\infty} E = \left\{ \left(\mathsf{j}_{\mu_{t}}^{\infty} s_{1} \right)'(0) \in T_{\theta} \mathsf{J}^{\infty} E \mid \mu \in \mathbb{C}^{\infty} \left(I_{\varepsilon}, M \right) \& \mu_{0} = p \right\} \\ = \left\{ \left(\mathsf{j}_{\mu_{t}}^{\infty} s_{2} \right)'(0) \in T_{\theta} \mathsf{J}^{\infty} E \mid \mu \in \mathbb{C}^{\infty} \left(I_{\varepsilon}, M \right) \& \mu_{0} = p \right\}.$$

$$(1.1.5)$$

1.1.5 Remark The lemma implies in particular that the horizontal space $C_{\theta}J^{\infty}E$ does not depend on the choice of a section representing θ .

Proof. After possibly shrinking U and ε choose a trivialising fibered chart $(x, u) : \widetilde{U} \to \mathbb{R}^d \times \mathbb{R}^n$ around an open neighborhood of $s_1(p) = s_2(p)$ as above. Moreover, we can assume after possible shrinking U and ε again that both $s_1(U)$ and $s_2(U)$ are contained in \widetilde{U} . Then compute

$$(\mathbf{j}_{\mu t}^{\infty} s_{1})'(0) = \sum_{i=1}^{d} (\mu^{i})'(0) \left(\frac{\partial}{\partial x^{i}} + \sum_{a=1}^{n} \sum_{\mathbf{I}} \frac{\partial^{|I|+1} s_{1}^{a}}{\partial x^{i} \partial x^{\mathbf{I}}} (p) \frac{\partial}{\partial u_{\mathbf{I}}^{a}} \right)$$

$$= \sum_{i=1}^{d} (\mu^{i})'(0) \left(\frac{\partial}{\partial x^{i}} + \sum_{a=1}^{n} \sum_{\mathbf{I}} \frac{\partial^{|I|+1} s_{2}^{a}}{\partial x^{i} \partial x^{\mathbf{I}}} (p) \frac{\partial}{\partial u_{\mathbf{I}}^{a}} \right) = (\mathbf{j}_{\mu t}^{\infty} s_{2})'(0)$$

where $\mu^i = x^i \circ \mu$, $s_j^a = u^a \circ s_j$ for j = 1, 2, and where I runs through the Roman multi-indices in the index set J. This proves the claim.

1.1.6 Lemma For every section $s \in \mathcal{E}_U$ the map

$$T_p M \to T_p M, \ \mu'(0) \mapsto \left(\pi_\infty \circ \mathsf{j}_{\mu_t}^\infty(s)\right)'(0)$$

is the identity map, where tangent vectors at p are represented as derivatives at the base point 0 of smooth paths $\mu : I_{\varepsilon} \to M$ based at p that is which fulfill $\mu_0 = p$.

Proof. This is trivial, since $\pi_{\infty} \circ j_{\mu_t}^{\infty}(s) = \mu_t$ for all $t \in I_{\varepsilon}$.

Despite the lemma being trivial, some of its consequences are not.

1.1.7 Proposition For every smooth fiber bundle $\pi : E \to M$ the Cartan distribution is a smooth involutive vector subbundle of the tangent bundle on $J^{\infty}E$. The Cartan distribution has fiber dimension $d = \dim M$. In a fibered chart $(x, u) : \tilde{U} \to \mathbb{R}^d \times \mathbb{R}^n$, a local frame for the Cartan distribution is given by the family of vector fields

$$D_i = \frac{\partial}{\partial x^i} + \sum_{a=1}^n \sum_{I} u_{Ii}^a \frac{\partial}{\partial u_{I}^a} , \quad i = 1, \dots, d ,$$

where the right summation is taken over all Roman multi-indices I in the index set $\mathcal{I} = \{1, \ldots, d\}$.

Proof. By Lemma 1.1.6 it is clear that dim $C_{\theta} J^{\infty} E = d$ for every $\theta \in J^{\infty} E$.

1.2. Euler-Lagrange equations

Regular domains

Before we come to the Euler-Lagrange equations of a variational problem we need to explain the kind of domains over which we want to consider variational problems. To this end recall first that by a triangulation of a topological space X one understands a homeomorphism of the form

$$\kappa : |K| \to X$$
,

where |K| is the underlying topological space of a (geometric) simplicial complex in some euclidean space \mathbb{R}^n . In the case where the topologial space X is a closed subset of a compact manifold-withboundary \overline{M} we call the triangulation *piecewise smooth* if for every simplex $\sigma \in K$ the restriction $\kappa|_{\sigma} : \sigma \to \kappa(\sigma)$ is a diffeomorphism onto its image which means that the following two conditions hold.

- (i) For every smooth f defined on an open neighborhood of $\kappa(\sigma)$, the pullback $(\kappa|_{\sigma})^* f$ can be extended to a smooth function on the euclidean space \mathbb{R}^n in which the simplicial complex K lies.
- (ii) For every smooth g defined on an open neighborhood of the simplex $\sigma \subset \mathbb{R}^n$, the pullback $(\kappa|_{\sigma}^{-1})^*g$ has a smooth extension to \overline{M} .

After these preparatory remarks let M be a smooth manifold of finite type that is let M be diffeomorphic to the interior of a compact manifold-with-boundary \overline{M} . Denote by d the dimension of Mand assume that d > 0. By a *regular domain* in M we now understand a non-empty open connected subset $\Omega \subset M$ such that its closure $\overline{\Omega}$ in \overline{M} possesses a piecewise smooth triangulation $\kappa : |K| \to \overline{\Omega}$, where |K| is the underlying topological space of a finite (geometric) simplicial complex. It is further assumed that $\kappa^{-1}(\partial\Omega)$ and $\kappa^{-1}(\partial\Omega \cap \partial M)$ are simplicial subcomplex of K of dimension < d where $\partial\Omega$ denotes the topological boundary of Ω in \overline{M} and ∂M is the boundary $\overline{M} \backslash M$.

Regular domains will comprise the domains over which we consider variational problems. In most applications, Ω will be the interior of a submanifold-with-corners of \overline{M} ; see MR [1] and Joyce (2012) for details on manifolds-with-corners and their submanifolds. In this section we will therefore consider mostly this particular case and only briefly indicate how the argument goes in the more general situation. For ease of exposition we will call a regular domain $\Omega \subset M$ such that $\overline{\Omega} \subset M$ is a submanifold-with-corners a *strongly regular domain*. If additionally the boundary $\partial\Omega$ is even smooth, then we call Ω a *strongly regular domain with smooth boundary*.

In most cases we choose M to coincide with the euclidean space \mathbb{R}^d . Note that \mathbb{R}^d is a manifold of finite type and that it is diffeomorphic to the interior of the d-dimensional closed unit ball $\overline{\mathbb{B}}^d$ around the origin. A diffeomorphism between the euclidean space \mathbb{R}^d and the interior \mathbb{B}^d of $\overline{\mathbb{B}}^d$ is given by the smooth map $\varphi : \mathbb{R}^d \to \mathbb{B}^d$, $x \mapsto \frac{1}{\sqrt{1+\|x\|^2}} x$. It has inverse $\psi : \mathbb{B}^d \to \mathbb{R}^d$, $y \mapsto \frac{1}{\sqrt{1-\|y\|^2}} y$ as the following two equalities show.

$$\begin{split} \varphi(\psi(y)) &= \frac{1}{\sqrt{1 + \frac{\|y\|^2}{1 - \|y\|^2}}} \frac{1}{\sqrt{1 - \|y\|^2}} y = y \\ \psi(\varphi(x)) &= \frac{1}{\sqrt{1 - \frac{\|x\|^2}{1 + \|x\|^2}}} \frac{1}{\sqrt{1 + \|x\|^2}} x = x \end{split}$$

In the remainder of this section we will identify \mathbb{R}^d with its image in the *d*-dimensional closed ball $\overline{\mathbb{B}}^d$. Under this identification one can understand $\overline{\mathbb{B}}^d$ as a certain compactification of euclidean space \mathbb{R}^d . It is termed the *radial compactification* of \mathbb{R}^d and sometimes denoted by $\overline{\mathbb{B}}^d_{\infty}$. Last, we call the boundary $\partial \overline{\mathbb{B}}^d_{\infty} = \overline{\mathbb{B}}^d_{\infty} \setminus \varphi(\mathbb{R}^d)$ the (d-1)-sphere at infinity and denote it by the symbol $\mathbb{S}^{d-1}_{\infty}$.

The local case

1.2.1 Assume that $\Omega \subset \mathbb{R}^d$ is an open subset which can be identified with the interior of a compact submanifold-with-boundary $\overline{\Omega} \subset \overline{\mathbb{B}}^d_{\infty}$ under the above identification $\varphi : \mathbb{R}^d \to \mathbb{B}^d_{\infty}$. In particular this means that the boundary $\partial\Omega$ is a closed submanifold of $\overline{\mathbb{B}}^d_{\infty}$. Later we will relax the assumptions and allow Ω to be a regular domain in \mathbb{R}^d . We interpret the preimages $\varphi^{-1}(\overline{\Omega})$ and $\varphi^{-1}(\partial\Omega)$ as intersections $\overline{\Omega} \cap \mathbb{R}^d$ and $\partial\Omega \cap \mathbb{R}^d$, respectively. Note that both of these spaces are submanifolds of \mathbb{R}^d , the first one possibly with boundary. Let $(x^1, \ldots, x^d) : \Omega \to \mathbb{R}^d$ be the canonical coordinates of Ω . Observe that $\overline{\Omega} \cap \mathbb{R}^d$ and Ω are oriented by the restriction of the canonical volume form $dx^1 \wedge \ldots \wedge dx^d$ to $\overline{\Omega} \cap \mathbb{R}^d$. We denote that restriction by ω .

Further we assume to be given a trivial smooth fiber bundle $\pi : E = \overline{\Omega} \times F \to \overline{\Omega}$ with typical fiber F being a connected open subset of some euclidean space \mathbb{R}^n . The canonical fiber coordinates will

be denoted by $(u^1, \ldots, u^n) : F \to \mathbb{R}^n$. The canonical charts of the interior of the base and the fiber give rise to a fibered chart $(x, u) : E = \Omega \times F \to \mathbb{R}^d \times \mathbb{R}^n$. The fiber bundle $\pi : E \to \overline{\Omega}$ and its associated jet bundles give rise to various kinds of section spaces which we need in the following and which we briefly now recall. Assume to be given some order $m \in \mathbb{N} \cup \{\infty\}$ and a locally closed subset $X \subset \overline{\Omega}$. Let $\tilde{E} \to \overline{\Omega}$ be one of the bundles $E \to \overline{\Omega}$ or $\mathsf{J}^k E \to \overline{\Omega}$, where $k \in \mathbb{N} \cup \{\infty\}$. By $\Gamma^m(X; \tilde{E})$ we then denote the space of *m*-times continuously differentiable sections of \tilde{E} over X that is of all continuous sections $s : X \to \tilde{E}$ which have an *m*-times continuously differentiable extension to an open neighborhood of X in $\overline{\Omega}$. The subspace of all $s \in \Gamma^m(X; \tilde{E})$ with support compactly contained in $X \cap \mathbb{R}^d$ will be denoted by $\Gamma_0^m(X; \tilde{E})$. We often write $\Gamma(X; \tilde{E})$ instead of $\Gamma^0(X; \tilde{E})$ for the space of continuous sections. The space $\mathcal{E}^m(X; E)$ of *Whitney fields* of order *m* over X with values in *E* is defined by

$$\mathcal{E}^m(X;E) = \left\{ S \in \Gamma(X;\mathsf{J}^m E) \mid \exists s \in \Gamma^m(\bar{\Omega};E) : \, \mathsf{j}^m s |_X = S \right\} \,.$$

Analogously as for Γ^m we denote by $\mathcal{E}_0^m(X; E)$ the space of all Whitney fields $S \in \mathcal{E}^m(X; E)$ which have support compactly contained in $X \cap \mathbb{R}^d$. Note that each of the section spaces $\Gamma^m(X; \tilde{E})$ and $\mathcal{E}^m(X; E)$ can be written as the quotient of some function space $\mathcal{C}^m(U, F \times \mathbb{R}^l)$, where $l \in \mathbb{N} \cup \{\infty\}$ and $U \subset \overline{\Omega}$ is open. Therefore, each of those sections spaces inherits from the corresponding $\mathcal{C}^m(U, F \times \mathbb{R}^l)$ the structure of a Fréchet space. As a consequence of this observation, the section spaces $\Gamma_0^m(X; \tilde{E})$ and $\mathcal{E}_0^m(X; E)$ become LF-spaces in a natural way.

The next ingredient we need is a *lagrangian function* that is a function $L \in C^{\infty}_{loc}(J^{\infty}\pi)$. Since L is a local function on the jet bundle, it can be regarded as an element of $C^{\infty}(J^{k}\pi)$ for some natural k. Let $\operatorname{ord}(L)$ be the smallest of such numbers and call it the *order* of the langragian function. The canonical volume form ω together with the lagrangian L give rise to the *lagrangian density* $\mathcal{L} = L \omega$ on the jet bundle $J^{\infty}\pi$. todo: add normalized lagrangians

Before we can write down the action functional induced by the lagrangian density \mathcal{L} we need to fix some boundary conditions. For now, we will restrict to *Cauchy boundary conditions with compact support* of some given order $m \in \mathbb{N} \cup \{\infty\}$. These are encoded by Whitney fields $F \in \mathcal{E}^m(\partial\Omega; E) \subset \Gamma(\partial\Omega; J^m E)$ with support being compact and contained in $\partial\Omega \cap \mathbb{R}^d$. More precisely, define the space of *Cauchy* boundary data of order m over the regular domain Ω by

$$\begin{aligned} \mathcal{B}^m_{\mathsf{Cauchy}}(\partial\Omega; E) &:= \mathcal{E}^m(\partial\Omega; E) \cap \Gamma_0^\infty(\partial\Omega; \mathsf{J}^m E) = \\ &= \left\{ B \in \Gamma^\infty(\partial\Omega; \mathsf{J}^m E) \; \middle| \; \exists b \in \Gamma^\infty(\bar{\Omega}; E) : \mathsf{j}^m b |_{\partial\Omega} = B \, \& \; \mathrm{supp} \, b \Subset \bar{\Omega} \cap \mathbb{R}^d \right\} \,. \end{aligned}$$

Given an element $B \in \mathcal{B}^m_{Cauchy}(\partial\Omega; E)$, we single out the space X_B of allowable sections of E:

$$\mathbf{X}_B = \left\{ s \in \Gamma_0^\infty(\bar{\Omega}; E) \mid \mathbf{j}^m s |_{\partial \Omega} = B \right\} \,.$$

In other words, X_B consists of all smooth sections $s : \overline{\Omega} \to E$ which fulfill the support condition $\sup s \Subset \overline{\Omega} \cap \mathbb{R}^d$ and the Cauchy boundary condition $j^m s|_{\partial\Omega} = B$. Observe that by construction X_B is an affine space over the vector space

$$\mathbf{V}^m = \left\{ v \in \Gamma_0^\infty(\bar{\Omega}; E) \mid \mathbf{j}^m v |_{\partial \Omega} = 0 \right\} \,.$$

That space carries a natural locally convex topology given by the locally convex colimit topology of the strict inductive system of Fréchet spaces

$$\mathbf{V}_N^m = \left\{ v \in \mathbf{V}^m \mid \operatorname{supp} v \subset \overline{\Omega} \cap \overline{\mathbb{B}}_N(0, \mathbb{R}^d) \right\}, \quad N \in \mathbb{N}.$$

The affine space X_B inherits the locally convex topology from V^m and thus becomes a manifold globally modeled on V^m . The tangent bundle of X_B then is canonically isomorphic to the product manifold $X_B \times V^m$.

Now we can write down the *action functional* associated to the lagrangian density \mathcal{L} :

$$S: X_B \to \mathbb{R}, \ s \mapsto \int_{\bar{\Omega} \cap \mathbb{R}^d} (j^{\infty} s)^* \mathcal{L} = \int_{\bar{\Omega} \cap \mathbb{R}^d} (L \circ j^{\infty} s) \cdot \omega \ . \tag{1.2.1}$$

Note that even though the domain of integration might be unbounded, the integral is well-defined for every $s \in X_B$ since $L \circ j^{\infty}s$ has compact support contained in $\overline{\Omega} \cap \mathbb{R}^d$ whenever s has that property.

1.2.2 Proposition Assume that $\Omega \subset \mathbb{R}^d$ is an open subset such that its closure $\overline{\Omega}$ in $\overline{\mathbb{B}}^d_{\infty}$ is a submanifold-with-boundary. Denote by ω the canonical volume element on Ω induced from \mathbb{R}^d . Assume further that $\pi : E = \overline{\Omega} \times F \to \overline{\Omega}$ a trivial fiber bundle with typical fiber F being an open and connected subset of some \mathbb{R}^n . Let $L \in C^{\infty}_{loc}(J^{\infty}\pi)$ be a lagrangian, and B an element of the space $\mathcal{B}^m_{Cauchy}(\Omega; E)$ of Cauchy boundary data of order $m \ge \operatorname{ord}(L) - 1$. Then the action functional $S : X_B \to \mathbb{R}$ associated to the lagrangian density $\mathcal{L} = L\omega$ is continuous. Moreover, S is Gateaux differentiable. The corresponding functional derivative $\delta S : TX_B = X_B \times V^m \to \mathbb{R}$ is linear in V^m , continuous and given by

$$\left\langle \delta \,\mathbf{S}(s), v \right\rangle = \sum_{\mathbf{a}=1}^{n} \int_{\bar{\Omega} \cap \mathbb{R}^{d}} v^{\mathbf{a}} \cdot \mathbf{j}^{\infty}(s)^{*} \left(\frac{\partial L}{\partial u^{\mathbf{a}}} + \sum_{\mathbf{I} \in \bar{\mathcal{I}}^{\bullet}, |\mathbf{I}| > 0} (-1)^{|I|} \frac{\partial^{|\mathbf{I}|}}{\partial x^{\mathbf{I}}} \left(\frac{\partial L}{\partial u^{\mathbf{a}}_{\mathbf{I}}} \right) \right) \cdot \omega , \qquad (1.2.2)$$

where $(s, v) \in X_B \times V^m$ and where v^a is the composition $u^a \circ v$.

Proof. We first show that the functional S is sequentially continuous with respect to the locally convex topology on X_B which means that for each $s \in X_B$ and each sequence $(s_k)_{k \in \mathbb{N}}$ in X_B converging to s the sequence $(S(s_k))_{k \in \mathbb{N}}$ converges to S(s). Since V^m is an LF space that is the locally convex colimit of a countable strict inductive system of Fréchet spaces, there exists a positive natural number N such that $s_k - S \in V_N^m$ for all $k \in \mathbb{N}$. Since the support of s is compactly contained in $\overline{\Omega} \cap \mathbb{R}^d$, we can assume after possibly increasing N that $\operatorname{supp} s \subset \overline{\mathbb{B}}_N(0, \mathbb{R}^d)$. Hence the supports all s_k are contained in $\overline{\Omega} \cap \overline{\mathbb{B}}_N(0, \mathbb{R}^d)$, and for every $\alpha \in \mathbb{N}^d$ the sequence $\left(\frac{\partial^{|\alpha|}s_k}{\partial x^\alpha}\right)_{k \in \mathbb{N}}$ converges uniformly on $\overline{\Omega} \cap \overline{\mathbb{B}}_N(0, \mathbb{R}^d)$ to $\frac{\partial^{|\alpha|}s}{\partial x^\alpha}$. Since the lagrangian function L has finite order, the compositions $L \circ j^\infty s_k$ and $L \circ j^\infty s$ also have compact support contained in $\overline{\Omega} \cap \overline{\mathbb{B}}_N(0, \mathbb{R}^d)$ to $L \circ j^\infty s$. Hence the sequence of integrals $\int_{\overline{\Omega} \cap \mathbb{R}^d} (L \circ j^\infty s_k) \omega$ converges to $S(s) = \int_{\overline{\Omega} \cap \mathbb{R}^d} (L \circ j^\infty s) \omega$, and the action functional is sequentially continuous.

The proof of sequential continuity can not be extended to also show continuity of the action just by replacing sequences with nets. The reason is that a net in an LF space, e.g. one labeled by the first uncountable ordinal, might not have any subsequences at all. Hence, unlike a converging sequences, a net in an LF space need not eventually be contained in one of the Fréchet spaces of the strict inductive system defining the LF structure. This observation makes the main ingredient in the above argument fail for the case of converging nets. One therefore needs another approach to prove continuity of S. We will use the observation from MR [2] that the locally convex topology of the LF space V^m is defined by the collection of all seminorms

$$p_{N,\theta}: \mathbf{V}^m \to \mathbb{R}_{\geq 0}, \ v \mapsto \sup_{1 \leq a \leq n} \sup_{\mathbf{I} \in \bar{\jmath}^{\bullet}, |\mathbf{I}| \leq N} \left\| \theta_I \frac{\partial^{|\mathbf{I}|} v^a}{\partial x^{\mathbf{I}}} \right\|_{\bar{\Omega}}$$

Let $s \in X_B$ and choose $N \in \mathbb{N}$ large enough so that $\operatorname{supp} s \Subset \overline{\Omega} \cap \mathbb{B}_N(0, \mathbb{R}^d)$.

Next we prove Gateaux differentiability. To this end let $s_{\bullet} = (s_t)_{t \in I_{\varepsilon}}$ be a smooth path in X_B defined on some open interval I_{ε} of the form $(-\varepsilon, \varepsilon)$ with $\varepsilon > 0$ and which fulfills $s_0 = s$. Note that the tangent vector \dot{s}_0 is an element of the space V^m and that every $v \in V$ can be obtained as the tangent vector of a smooth path in X_B , e.g. of the path $\mathbb{R} \to X_B$, $t \mapsto s + tv$. Denote by o the order of the langrangian and by \mathcal{I} the index set $\{1, \ldots, d\}$. Recall Equation 1.1.3 for the vertical derivative of the jet map:

$$T\mathbf{j}^{\infty}(\dot{s}_{0}) = \left. \frac{d}{dt} \mathbf{j}^{\infty} s_{t} \right|_{t=0} = \sum_{\mathbf{a}=1}^{n} \sum_{\mathbf{I} \in \overline{\mathbf{J}}^{\bullet}} \frac{\partial^{|\mathbf{I}|} \dot{s}_{0}^{\mathbf{a}}}{\partial x^{\mathbf{I}}} \frac{\partial}{\partial u_{\mathbf{I}}^{\mathbf{a}}} .$$
(1.2.3)

In this formula, $\overline{\mathfrak{I}}^{\bullet}$ denotes the set of *ordered Roman multi-indices* in the set $\mathfrak{I} = \{1, \ldots, d\}$ that is $\overline{\mathfrak{I}}^{\bullet}$ consists of a zero element O and all finite sequences of integers of the form

$$I = (i_1, \dots, i_k)$$
, where $k \in \mathbb{N}_{>0}$ and $1 \leq i_1 \leq i_2 \leq \dots \leq i_k \leq d$.

The number k is called the *length* of the ordered Roman multi-index I. The *length* of the zero element O is defined to be 0. See Section A.8.1 for details on Roman multi-indices and their combinatorial properties. After these preparations we now compute:

$$\frac{d}{dt} \mathbf{S}(s_t) \Big|_{t=0} = \int_{\bar{\Omega} \cap \mathbb{R}^d} \frac{d}{dt} L \circ \mathbf{j}^{\infty}(s_t) \Big|_{t=0} \omega = \sum_{\mathbf{a}=1}^n \int_{\bar{\Omega} \cap \mathbb{R}^d} \sum_{\mathbf{I} \in \bar{\mathbf{J}}^{\bullet}} \frac{\partial^{|\mathbf{I}|} \dot{s}_0^{\mathbf{a}}}{\partial x^{\mathbf{I}}} \left(\frac{\partial L}{\partial u_{\mathbf{I}}^{\mathbf{a}}} \circ \mathbf{j}^{\infty}(s_0) \right) \omega = \\ = \sum_{\mathbf{a}=1}^n \int_{\bar{\Omega} \cap \mathbb{R}^d} \left(\dot{s}_0^{\mathbf{a}} \left(\frac{\partial L}{\partial u^{\mathbf{a}}} \circ \mathbf{j}^{\infty}(s_0) \right) + \sum_{\mathbf{I} \in \bar{\mathbf{J}}^{\bullet}, \, |\mathbf{I}| > 0} \frac{\partial^{|\mathbf{I}|} \dot{s}_0^{\mathbf{a}}}{\partial x^{\mathbf{I}}} \left(\frac{\partial L}{\partial u_{\mathbf{I}}^{\mathbf{a}}} \circ \mathbf{j}^{\infty}(s_0) \right) \right) \omega .$$

1.2.3 We now want to find the extremal points of the functional S, if such exist. To this end we first derive a necessary condition for $s_0 \in X_B$ to be an extremal point of S.

I.2. Semi-riemannian geometry

2.1. Causal structures

2.1.1 In this section, we let (M, g) denote a connected lorentzian manifold of dimension D = d + 1, $d \in \mathbb{N}_{>0}$. In particular this means that the signature of the semi-riemannian structure g is (1, d) or, in different notation, $(+, -, \ldots, -)$. At each point $p \in M$ the tangent space T_pM then canonically carries the structure of a D-dimensional lorentzian vector space. Denote by $q_{\mathsf{L}}: TM \to \mathbb{R}$ the *Lorentz quadratic form* $v \mapsto g(v, v)$. With these notational agreements in mind we now make the following definition.

2.1.2 Definition A tangent vector $v \in TM$ is called

- (i) *lightlike* or *null* if $v \neq 0$ and $q_{\mathsf{L}}(v) = 0$,
- (ii) timelike if $q_{\mathsf{L}}(v) > 0$,
- (iii) spacelike if v = 0 or $q_{L}(v) < 0$, and
- (iv) causal (or non-spacelike) if $v \neq 0$ and $q_{L}(v) \ge 0$.

A piecewise differentiable curve $\gamma : [a, b] \to M$, $-\infty \leq a < b \leq \infty$, is called *lightlike*, *timelike*, or *spacelike* if each of its tangent vectors is so, respectively.

2.1.3 Proposition

2.1.4

From now on, we assume that M is *temporally orientable* that is that

Part II. Quantum Mechanics

II.1. The postulates of quantum mechanics

1.1. The geometry of projective Hilbert spaces

1.1.1 Let \mathcal{H} be a Hilbert space over the field $\mathbb{K} = \mathbb{R}$ or $= \mathbb{C}$. The associated *projective Hilbert space* $\mathbb{P}\mathcal{H}$ then is defined as the space of all *rays* in \mathcal{H} that is as the space

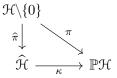
 $\mathbb{PH} = \left\{ \ell \in \mathcal{P}(\mathcal{H}) \mid \ell \text{ is a 1-dimensional } \mathbb{K}\text{-linear subspace of } \mathcal{H} \right\} \,.$

It carries a natural topology which we now describe. Consider $\mathcal{H}\setminus\{0\}$ with its subspace topology. Then one has a natural map

$$\pi: \mathfrak{H} \setminus \{0\} \to \mathbb{P} \mathfrak{H}, v \mapsto \mathbb{K} v$$

which obviously is surjective. One endows \mathbb{PH} with the final topology with respect to π . Next let us introduce an equivalence relation \sim on $\mathcal{H}\setminus\{0\}$ by defining $v \sim w$ if there exists a $\lambda \in \mathbb{K}^{\times} = \mathbb{K}\setminus\{0\}$ such that $v = \lambda w$. Obviously \sim is reflexive, since $1 \in \mathbb{K}^{\times}$, symmetric, since with $\lambda \in \mathbb{K}^{\times}$ the inverse λ^{-1} is in \mathbb{K}^{\times} as well, and transitive, since the product of two elements of \mathbb{K}^{\times} is in \mathbb{K}^{\times} . Hence \sim is an equivalence relation indeed. Denote by \hat{v} the equivalence class of an element $v \in \mathcal{H}\setminus\{0\}$. Let $\hat{\mathcal{H}}$ be the quotient space $(\mathcal{H}\setminus\{0\})/\sim$ and $\hat{\pi}: \mathcal{H}\setminus\{0\} \to \hat{\mathcal{H}}$ the quotient map.

1.1.2 Lemma The map $\pi : \mathfrak{H} \setminus \{0\} \to \mathbb{P} \mathfrak{H}$ factors through a unique homeomorphism $\kappa : \hat{\mathfrak{H}} \to \mathbb{P} \mathfrak{H}$ which means that the diagram



commutes and that κ is uniquely determined by this condition.

Proof. If $v \sim w$, then the lines through v and through w coincide, hence π factors through a unique continuous map $\kappa : \hat{\mathcal{H}} \to \mathbb{P}\mathcal{H}$ by the universal property of the quotient space. By surjectivity of π , κ is surjective, too. By definition, κ maps \hat{v} to $\mathbb{K}v$, hence if $\mathbb{K}v = \mathbb{K}w$, then v and w are linearly dependant, and $v \sim w$. So κ is injective. Continuity of the inverse $\kappa^{-1} : \mathbb{P}\mathcal{H} \to \hat{\mathcal{H}}$ is a consequence of the fact that $\mathbb{P}\mathcal{H}$ carries the final topology with respect to π . Uniqueness of κ follows from $\hat{\pi}$ being surjective.

1.1.3 Lemma The projection map $\pi : \mathcal{H} \setminus \{0\} \to \mathbb{P}\mathcal{H}$ and its restriction $\pi|_{\mathbb{S}\mathcal{H}} : \mathbb{S}\mathcal{H} \to \mathbb{P}\mathcal{H}$ to the sphere of \mathcal{H} are open.

Proof. By the preceding lemma it suffices to show that $\hat{\pi} : \mathcal{H} \setminus \{0\} \to \hat{\mathcal{H}}$ is open. Let $U \subset \mathcal{H} \setminus \{0\}$ be open. Then

$$\widehat{\pi}^{-1}(\widehat{\pi}(U)) = \bigcup_{\lambda \in \mathbb{K}^{\times}} \lambda \cdot U ,$$

which is again open and the first part of the claim is proved. The second part follows in the same way, since

$$\widehat{\pi}|_{\mathbb{SH}}^{-1}(\widehat{\pi}|_{\mathbb{SH}}(U)) = \bigcup_{\lambda \in \mathbb{S}(\mathbb{K})} \lambda \cdot U$$

is open for all $U \subset \mathbb{SH}$ open.

1.1.4 Remark Strictly speaking, the projective space \mathbb{PH} depends on the ground field \mathbb{K} . If \mathcal{H} is a complex Hilbert space one therefore sometimes writes \mathbb{RPH} or \mathbb{CPH} to denote that the projective space of all real respectively all complex lines is meant. In this work we agree that for \mathcal{H} complex \mathbb{PH} always stands for the projective space of complex lines in \mathcal{H} . If we want to consider the projective space of real lines in some complex Hilbert space \mathcal{H} instead, we write \mathbb{RPH} .

1.1.5 The inner product on the underlying Hilbert space \mathcal{H} induces the *projective inner product* or *ray inner product*

$$\langle\!\langle\cdot,\cdot\rangle\!\rangle: \mathbb{PH} \times \mathbb{PH} \to [0,1], \ (\mathbb{K}v, \mathbb{K}w) \mapsto \langle\!(\mathbb{K}v, \mathbb{K}w)\!\rangle = \frac{|\langle v, w\rangle|}{\|v\| \|w\|}, \quad \text{where } v, w \in \mathcal{H} \setminus \{0\},$$

on the associated projective space. Note that the projective inner product is well-defined, since $\frac{\langle v, w \rangle}{\|v\| \|w\|}$ is homogeneous of degree 0 both in v and w.

Now we can formulate the first postulate of quantum mechanics.

(QM1) The state space of a quantum mechanical system is accomplished by a projective space \mathbb{PH} associated to a complex separable Hilbert space \mathcal{H} . The elements $v \in \mathcal{H} \setminus \{0\}$ are called state vectors, the rays $\ell \in \mathbb{PH}$ are the *pure states*.

If a quantum mechanical system is prepared so that it is in the state $\ell \in \mathbb{PH}$, the probability that a measurement detects the system to be in the state $k \in \mathbb{PH}$ is given by the *transition* probability $(k, \ell)^2$.

Because of their appearance in the first postulate of quantum mechanics we want to study projective Hilbert spaces in some more depth. We will use topological, geometric and analytic tools for that endeavor. A first result is the following.

1.1.6 Theorem Let \mathbb{PH} be the projective space of a Hilbert space of dimension ≥ 2 over the field \mathbb{K} of real or complex numbers. Then the following holds true:

- (i) The projective Hilbert space \mathbb{PH} is a completely metrizable topological space.
- (ii) A complete metric inducing the topology on \mathbb{PH} is given by

$$d: \mathbb{PH} \times \mathbb{PH} \to \mathbb{R}_{\geq 0}, \ (\mathscr{k}, \mathscr{\ell}) \mapsto \inf \left\{ \|v - w\| \mid v \in \mathscr{k}, \ w \in \mathscr{\ell} \& \|v\| = \|w\| = 1 \right\}$$

(iii) The metric d and the transition amplitudes satisfy the relation

$$d^{2}(\mathcal{K}, \ell) = 2(1 - \langle \mathcal{K}, \ell \rangle) \ge 1 - \langle \mathcal{K}, \ell \rangle^{2} \quad \text{for all } \mathcal{K}, \ell \in \mathbb{PH} .$$

$$(1.1.1)$$

(iv) *The* Fubini–Study distance

$$d_{\mathsf{FS}}: \mathbb{PH} \times \mathbb{PH} \to \mathbb{R}_{\geq 0}, \ (\pounds, \ell) \mapsto \arccos \langle \pounds, \ell \rangle$$

is a metric on \mathbb{PH} which is equivalent to the metric d. More precisely

$$d(\mathscr{k}, \mathscr{\ell}) \leqslant d_{\mathsf{FS}}(\mathscr{k}, \mathscr{\ell}) \leqslant \sqrt{2} d(\mathscr{k}, \mathscr{\ell}) \quad \text{for all } \mathscr{k}, \mathscr{\ell} \in \mathbb{PH} .$$

$$(1.1.2)$$

The diameter of \mathbb{PH} with respect to the Fubini–Study distance equals $\frac{\pi}{2}$.

(v) The mapping $P : \mathbb{PH} \to \mathfrak{B}(\mathfrak{H})$ which associates to every ray \mathscr{R} the orthogonal projection onto it is a bi-Lipschitz embedding. The gap metric

$$d_{\mathsf{gap}}: \mathbb{PH} \times \mathbb{PH} \to \mathbb{R}_{\geq 0}, \ (\mathscr{k}, \mathscr{\ell}) \mapsto \|P(\mathscr{k}) - P(\mathscr{\ell})\|$$

obtained by restricting the operator norm distance to \mathbb{PH} is equivalent to d and satisfies

$$\frac{1}{\sqrt{2}}d(\mathcal{k},\mathcal{\ell}) \leqslant d_{\mathsf{gap}}(\mathcal{k},\mathcal{\ell}) = \sqrt{1 - \langle \mathcal{k},\mathcal{\ell} \rangle^2} \leqslant d(\mathcal{k},\mathcal{\ell}) \quad \text{for all } \mathcal{k},\mathcal{\ell} \in \mathbb{PH} .$$
(1.1.3)

Proof. ad (ii)Let us first show that the map d is a metric indeed. By definition, d is non-negative and symmetric. Assume $d(\mathscr{K}, \mathscr{C}) = 0$ for two rays \mathscr{K}, \mathscr{C} . For given unit vectors $v \in \mathscr{K}$ and $w \in \mathscr{C}$ there then exists a sequence $(\sigma_k)_{k \in \mathbb{N}} \subset \mathbb{S}^1$ such that

$$\lim_{k\to\infty} \|v - \sigma_k w\| = 0 \; .$$

By compactness of \mathbb{S}^1 we can assume that the sequence $(\sigma_k)_{k\in\mathbb{N}}$ converges after possibly passing to a subsequence. Let $\sigma \in \mathbb{S}^1$ be its limit. Then $||v - \sigma w|| = 0$, hence $\mathscr{k} = \mathscr{C}$. Now let $\mathscr{K}, \mathscr{C}, \mathscr{J} \in \mathbb{PH}$ and $z \in \mathscr{J}$ a representing unit vector. Then

$$\begin{split} d(\&, \ell) &= \inf \left\{ \|v - w\| \mid v \in \&, \ w \in \ell \& \|v\| = \|w\| = 1 \right\} \leqslant \\ &\leq \inf \left\{ \|v - z\| + \|z - w\| \mid v \in \&, \ w \in \ell \& \|v\| = \|w\| = 1 \right\} = \\ &= \inf \left\{ \|v - z\| \mid v \in \& \& \|v\| = 1 \right\} + \inf \left\{ \|z - w\| \mid w \in \ell \& \|w\| = 1 \right\} = \\ &= d(\&, j) + d(j, \ell) \ , \end{split}$$

hence d satisfies the triangle inequality, and therefore is a metric.

Next we prove that the metric topology of d coincides with the quotient topology of π . Let $v, w \in SH$. By definition of the metric d one then has

$$d(\mathbb{K}v,\mathbb{K}w) \leq \|v-w\| .$$

This implies that for all $\varepsilon > 0$

$$\pi\left(\mathbb{B}_{\mathbb{S}\mathcal{H}}(v,\varepsilon)\right) \subset \mathbb{B}_{\mathbb{P}\mathcal{H}}(\mathbb{K}v,\varepsilon) ,$$

where $\mathbb{B}_{\mathbb{SH}}(v,\varepsilon)$ denotes the ε -ball around v in the sphere with respect to the norm and $\mathbb{B}_{\mathbb{PH}}(\mathbb{K}v,\varepsilon)$ the ε -ball around $\mathbb{K}v$ in the projective Hilbert space with respect to the metric d. Hence the quotient topology on \mathbb{PH} is finer than the metric topology. If for given $\varepsilon > 0$ a $\delta > 0$ is chosen so that $\delta < \varepsilon$, then for every ray ℓ with $d(\mathbb{K}v, \ell) < \delta$ there exists an element $w \in \ell \cap \mathbb{SH}$ such that $||v - w|| < \epsilon$ which means that $\ell = \pi(w) \in \pi(B(v, \varepsilon))$. Hence

$$\mathbb{B}_{\mathbb{P}\mathcal{H}}(\mathbb{K}v,\delta) \subset \pi\left(\mathbb{B}_{\mathbb{S}\mathcal{H}}(v,\varepsilon)\right)$$

and the quotient topology on \mathbb{PH} is coarser than the metric topology. So d induces the topology on \mathbb{PH} as claimed.

It remains to verify that d is a complete metric. To this end observe first that for every $v \in SH$ and ray ℓ there exists a representative $w \in \ell \cap SH$ such that $\langle v, w \rangle = \langle \mathbb{K}v, \ell \rangle$. We will call such a representative of ℓ distinguished with respect to v. Now let $(\ell_n)_{n \in \mathbb{N}}$ be a Cauchy sequence of rays. Then there exists an increasing sequence of natural numbers $n_0 < \ldots < n_k < n_{k+1} < \ldots$ such that

$$d(\ell_n,\ell_m) < \frac{1}{2^{k+1}} \quad \text{for all } n,m \geqslant n_k \ .$$

Choose a representative $v_0 \in \ell_{n_0} \cap S\mathcal{H}$ and let $v_1 \in S\mathcal{H}$ be a representative of ℓ_{n_1} distinguished with respect to v_0 . Then

$$\|v_1 - v_0\| = \sqrt{2(1 - \Re(v_0, v_1))} = \sqrt{2(1 - \langle \ell_{n_0}, \ell_{n_1} \rangle)} = d(\ell_{n_0}, \ell_{n_1}) < \frac{1}{2}.$$

Now assume we have constructed $v_0, \ldots, v_k \in SH$ such that $\mathbb{K}v_l = \ell_{n_l}$ for $l = 0, \ldots, k$ and such that for $l = 0, \ldots, k - 1$

$$\|v_{l+1} - v_l\| < \frac{1}{2^{l+1}}$$
 (1.1.4)

Let $v_{k+1} \in SH$ be a representative of $\ell_{n_{k+1}}$ distinguished with respect to v_k . Then

$$\|v_{k+1} - v_k\| = \sqrt{2(1 - \Re \mathfrak{e}\langle v_{k+1}, v_k \rangle)} = \sqrt{2(1 - \left|\langle \ell_{n_{k+1}}, \ell_{n_k} \right\rangle)} = d(\ell_{n_{k+1}}, \ell_{n_k}) < \frac{1}{2^{k+1}} .$$

We thus obtain a sequence $(v_k)_{k \in \mathbb{N}}$ in \mathcal{H} such that (1.1.4) is fulfilled for all $l \in \mathbb{N}$. The sequence $(v_k)_{k \in \mathbb{N}}$ is even a Cauchy sequence since for $n \ge m \ge k$

$$||v_n - v_m|| \leq \sum_{k=m}^{n-1} ||v_{k+1} - v_k|| < \sum_{k=m}^{n-1} \frac{1}{2^{k+1}} < \frac{1}{2^m}.$$

Let $v \in \mathcal{H}$ be its limit. Then

$$\lim_{k \to \infty} d(\mathbb{K}v, \mathscr{E}_{n_k}) \leq \lim_{k \to \infty} \|v - v_k\| = 0 .$$

Hence the sequence of rays $(\ell_n)_{n \in \mathbb{N}}$ converges to the ray $\mathbb{K}v$ and $\mathbb{P}\mathcal{H}$ is complete with respect to the metric d. Claim (i) is now proved as well.

ad (iii)Let &, ℓ be rays in \mathcal{H} and $v \in \&$, $w \in \ell$ representing unit vectors. Let $\lambda \in \mathbb{S}^1$ such that $\langle v, w \rangle = \lambda \langle \&, \ell \rangle$ and $\sigma \in \mathbb{S}^1$ arbitrary. Then compute

$$\|v - \sigma w\|^2 = 2(1 - \mathfrak{Re}\langle v, \sigma w \rangle) = 2(1 - \langle \mathfrak{K}, \ell \rangle) \mathfrak{Re} \,\overline{\sigma} \lambda) \ge 2(1 - \langle \mathfrak{K}, \ell \rangle) \ .$$

For $\sigma = \lambda$, equality holds, hence

$$d^{2}(\mathcal{K}, \mathcal{\ell}) = \inf \left\{ \|v - \sigma w\|^{2} \mid \sigma \in \mathbb{S}^{1} \right\} = 2\left(1 - |\langle \mathcal{K}, \mathcal{\ell} \rangle|\right).$$

With \mathcal{K}, ℓ, v, w as before and $\delta = d(\mathcal{K}, \ell)$, the claimed inequality now follows immediately:

$$d^{2}(\mathcal{R}, \mathcal{\ell}) \geq \delta^{2}\left(1 - \frac{1}{4}\delta^{2}\right) = 2\left(1 - \left|\left\langle \mathcal{R}, \mathcal{\ell}\right\rangle\right|\right)\left(1 - \frac{1}{2}\left(1 - \left|\left\langle \mathcal{R}, \mathcal{\ell}\right\rangle\right|\right)\right) = 1 - \left|\left\langle \mathcal{R}, \mathcal{\ell}\right\rangle\right|^{2} \right).$$

ad (iv)The map d_{FS} is symmetric by symmetry of the projective inner product. By the assumption $\dim \mathcal{H} \ge 2$, the image of $\langle \cdot, \cdot \rangle$ is the whole interval [0, 1], since $\mathbb{P}\mathcal{H}$ is connected, $\langle \cdot, \cdot \rangle$ is bounded by 1, $\langle \ell, \ell \rangle = 1$ for every ray ℓ and since there exist orthogonal rays. The image of d_{FS} therefore coincides with $[0, \frac{\pi}{2}]$ which already entails the claim about the diameter. By strict monotony of $\arccos d_{\text{FS}}(\mathscr{R}, \ell) = 0$ if and only if $\langle \mathscr{R}, \ell \rangle = 1$. By (1.1.1) this is the case if and only if $d(\mathscr{R}, \ell) = 0$ which means if and only if $\mathscr{R} = \ell$. Let us now show that d_{FS} satisfies the triangle inequality. To this end let $\mathscr{R}, \ell, \not{z}$ be rays in \mathcal{H} . If the Fubini–Study distance between any two of these rays is zero, the triangle inequality obviously holds true, so we exclude that case. Choose representatives $v \in \mathscr{R}$, $w \in \ell$, $z \in \mathcal{J}$ such that all have norm 1. After possibly multiplying v and z by elements of $\mathbb{S}^1 \cap \mathbb{K}$ one can achieve that

$$\langle v, w
angle = \langle \mathscr{k}, \mathscr{\ell}
angle$$
 and $\langle w, z
angle = \langle \mathscr{\ell}, \mathfrak{j}
angle$.

Let $\theta = \arccos\langle v, w \rangle$ and $\varphi = \arccos\langle w, z \rangle$. Then $\theta = d_{\mathsf{FS}}(\mathscr{R}, \mathscr{C})$ and $\varphi = d_{\mathsf{FS}}(\mathscr{C}, \mathscr{J})$. Now let x be a unit vector in the plane through v and w which is orthogonal to w and y a unit vector in the plane through w and z which is orthogonal to w. After possibly multiplying x and y by elements of $\mathbb{S}^1 \cap \mathbb{K}$ one can achieve that $\langle v, x \rangle, \langle z, y \rangle \in [0, 1]$. Then

$$v = \langle v, w \rangle w + \langle v, x \rangle x$$
 and $z = \langle z, w \rangle w + \langle z, y \rangle y$.

By $\theta, \varphi \in \left[0, \frac{\pi}{2}\right]$ and $\langle v, x \rangle, \langle z, y \rangle \ge 0$ one concludes

$$v = \cos \theta \, w + \sin \theta \, x$$
 and $z = \cos \varphi \, w + \sin \varphi \, y$

Hence, by the triangle inequality for the absolute value and the Cauchy-Schwarz inequality

$$|\langle v, z \rangle| = |\cos\theta \, \cos\varphi + \sin\theta \, \sin\varphi \, \langle x, y \rangle| \ge \cos\theta \, \cos\varphi - \sin\theta \, \sin\varphi = \cos(\theta + \varphi) \; .$$

Since arccos is monotone decreasing, one obtains

$$d_{\mathsf{FS}}(\mathscr{k}, \mathscr{j}) = \arccos |\langle v, z \rangle| \leqslant \theta + \varphi = d_{\mathsf{FS}}(\mathscr{k}, \mathscr{\ell}) + d_{\mathsf{FS}}(\mathscr{\ell}, \mathscr{j})$$

So the Fubini-Study distance satisfies the triangle inequality and is a metric indeed.

Last we need to prove that the Fubini–Study distance is equivalent to d. To this end consider the functions

$$f:[0,\sqrt{2}] \to \mathbb{R}, \ s \mapsto \arccos\left(1-\frac{s^2}{2}\right) \quad \text{and} \quad g:[0,\frac{\pi}{2}] \to \mathbb{R}, \ t \mapsto \sqrt{2(1-\cos t)} \ .$$

Then both functions are continuous and differentiable on the interior of their domains. Now observe that f(0) = g(0) = 0 and compute

$$f'(s) = \frac{s}{\sqrt{1 - \left(1 - \frac{s^2}{2}\right)^2}} = \frac{s}{\sqrt{s^2 - \frac{s^4}{4}}} = \frac{2}{\sqrt{4 - s^2}} \leqslant \sqrt{2} \quad \text{for } s \in (0, \sqrt{2})$$

and

$$g'(t) = \frac{\sqrt{2}}{2} \frac{\sin t}{\sqrt{1 - \cos t}} = \frac{\sqrt{2}}{2} \sqrt{1 + \cos t} \le 1 \quad \text{for } t \in (0, \frac{\pi}{2}) .$$

By definition of d_{FS} and (1.1.1), the mean-value theorem then entails

$$d(\mathscr{k}, \mathscr{\ell}) = g\left(d_{\mathsf{FS}}(\mathscr{k}, \mathscr{\ell})\right) \leqslant d_{\mathsf{FS}}(\mathscr{k}, \mathscr{\ell}) = f\left(d(\mathscr{k}, \mathscr{\ell})\right) \leqslant \sqrt{2} \, d(\mathscr{k}, \mathscr{\ell}) \quad \text{for all } \mathscr{k}, \mathscr{\ell} \in \mathbb{PH}$$

Hence the estimate (1.1.2) is proved and the metrics d and d_{FS} are equivalent.

ad (v)Recall that the operator norm distance of $P(\mathscr{R})$ and $P(\mathscr{C})$ is given by

$$\|P(\mathscr{K}) - P(\mathscr{\ell})\| = \sup\left\{ \left\| \left(P(\mathscr{K}) - P(\mathscr{\ell}) \right) z \right\| \mid z \in \mathbb{SH} \right\}$$
(1.1.5)

Choose normalized representatives $v \in \mathscr{K}$ and $w \in \mathscr{C}$. After possibly multiplying w by a complex number of modulus 1 we can assume that $\langle v, w \rangle = \langle \mathscr{K}, \mathscr{C} \rangle \ge 0$. If $\langle v, w \rangle = 1$ or in other words if v and w are linearly dependent then \mathscr{K} and \mathscr{C} coincide and the claim is trivial, so we assume that v and w are linearly independent. First we want to show that

$$\left\| \left(P(\mathscr{k}) - P(\mathscr{\ell}) \right) z \right\| \leq 1 - \left| \langle v, w \rangle \right|^2 \quad \text{for all } z \in \mathbb{SH} .$$
(1.1.6)

To this end expand $z = z^{\parallel} + z^{\perp}$, where z^{\parallel} lies in the plane spanned by v and w and z^{\perp} is perpendicular to that plane. Then

$$(P(\mathscr{k}) - P(\mathscr{\ell}))z = \langle z, v \rangle v - \langle z, w \rangle w = \langle z^{\parallel}, v \rangle v - \langle z^{\parallel}, w \rangle w = (P(\mathscr{k}) - P(\mathscr{\ell}))z^{\parallel}$$

Hence it suffices to verify (1.1.6) for $z \in SH \cap Span(v, w)$. Observe that there exist unique elements $\varphi \in [0, \frac{\pi}{2}]$ and $\mu \in S^1 \cap \mathbb{K}$ such that $\langle z, v \rangle = \overline{\mu} \cos \varphi$. One can then find a normalized vector $w^{\perp} \in Span(v, w)$ perpendicular to v such that

$$\mu z = \cos \varphi \, v + \sin \varphi \, w^{\perp} \; .$$

Note that with this

$$w = \langle v, w \rangle v + \langle w, w^{\perp} \rangle w^{\perp} \quad \text{and} \quad |\langle w, w^{\perp} \rangle|^2 = 1 - |\langle v, w \rangle|^2 \; .$$

Now compute

$$\begin{split} \left\| \left(P(\mathscr{R}) - P(\mathscr{C}) \right) z \right\|^2 &= \left\| \left(P(\mathscr{R}) - P(\mathscr{C}) \right) \mu z \right\|^2 = \left\| \langle z, v \rangle v - \langle z, w \rangle w \right\|^2 = \\ &= \left| \langle \mu z, v \rangle \right|^2 - 2 \langle v, w \rangle \operatorname{\mathfrak{Re}} \left(\langle \mu z, v \rangle \langle \mu z, w \rangle \right) + \left| \langle \mu z, w \rangle \right|^2 = \\ &= \cos^2 \varphi - 2 \cos \varphi \left\langle v, w \right\rangle \left(\cos \varphi \left\langle v, w \right\rangle + \sin \varphi \operatorname{\mathfrak{Re}} \langle w, w^\perp \rangle \right) \\ &+ \cos^2 \varphi \left| \langle v, w \rangle \right|^2 + 2 \cos \varphi \left\langle v, w \right\rangle \sin \varphi \operatorname{\mathfrak{Re}} \langle w, w^\perp \rangle + \sin^2 \varphi \left| \langle w, w^\perp \rangle \right|^2 = \\ &= 1 - \left| \langle v, w \rangle \right|^2 \,. \end{split}$$

This proves (1.1.6), but also implies by (1.1.5) that

$$\|P(\mathscr{R}) - P(\mathscr{C})\|^2 = 1 - |\langle v, w \rangle|^2 = 1 - |\langle \mathscr{R}, \mathscr{C} \rangle|^2$$

The claim now follows by (iii) and the theorem is proved.

After having examined some topological properties we come now to the geometry of projective Hilbert spaces.

1.1.7 Theorem The projective Hilbert space \mathbb{PH} of a Hilbert space of dimension ≥ 2 over \mathbb{K} has the following differential geometric properties:

 (i) PH carries a natural structure of an analytic manifold modelled on a Hilbert space isomorphic to each of the Hilbert spaces V_w = (Kw)[⊥], where w ∈ H is a unit vector.

(ii) Let $SH \subset H \setminus \{0\}$ be the sphere in H. Then the restriction

$$\pi|_{\mathbb{SH}}:\mathbb{SH}\to\mathbb{PH},\ v\mapsto\mathbb{K}v$$

is a real analytic fiber bundle with typical fiber \mathbb{S}^1 in the complex case and typical fiber $\mathbb{Z}/2$ in the real case.

- (iii) Endow SH with the riemannian metric g inherited from the ambient Hilbert space. Then there exists a unique riemannian metric g_{FS} on PH such that π|_{SH} : SH → PH becomes a riemannian submersion. This metric is called the Fubini–Study metric. Its geodesic distance coincides with the Fubini–Study distance d_{FS}.
- (iv) In case ℋ is a complex Hilbert space, the projective space ℙℋ carries in a natural way the structure of a Kähler manifold. Its complex structure is the one inherited from ℋ, and its riemannian metric is the Fubini–Study metric.

Proof. ad (i) For a given unit vector $w \in S\mathcal{H}$ consider the linear form $w^{\flat} : \mathcal{H} \to \mathbb{K}, v \mapsto \langle v, w \rangle$. Let $V_w = \ker w^{\flat} = (\mathbb{K}w)^{\perp}$ and $U_w = \pi(\mathcal{H} \setminus V_w)$. Then, by Theorem 3.2.3, one has the orthogonal decomposition $\mathcal{H} = V_w \oplus \mathbb{K}w$ which gives rise to the orthogonal projection $\mathrm{pr}_{V_w} : \mathcal{H} \to V_w$. Next observe that $U_w \subset \mathbb{P}\mathcal{H}$ is open since $\pi^{-1}(U_w) = \mathcal{H} \setminus V_w$ is open and $\mathbb{P}\mathcal{H}$ carries the quotient topology with respect to π . Now we can define a chart $h_w : U_w \to V_w$ by

$$h_w(\mathbb{K}v) = \operatorname{pr}_{\mathcal{V}_w}\left(\frac{v}{\langle v, w \rangle}\right) = \frac{v}{\langle v, w \rangle} - w \quad \text{for } v \in \mathcal{H} \setminus \mathcal{V}_w \;.$$

The map h_w is well-defined since $\langle v, w \rangle \neq 0$ for all $v \in \mathcal{H} \setminus V_w$ and since $\frac{v}{\langle v, w \rangle} = \frac{\lambda v}{\langle \lambda v, w \rangle}$ for all $\lambda \in \mathbb{K}^{\times}$. Moreover, h_w is continuous by continuity of the composition $h_w \circ \pi|_{\mathcal{H} \setminus V_w}$. If $h_w(\mathbb{K}v) = h_w(\mathbb{K}v')$, then

$$\mathrm{pr}_{\mathcal{V}_w}\left(\frac{v}{\langle v,w\rangle}-\frac{v'}{\langle v',w\rangle}\right)=0 \quad \text{and} \quad \left\langle \frac{v}{\langle v,w\rangle}-\frac{v'}{\langle v',w\rangle},w\right\rangle=0,$$

hence $\mathbb{K}v = \mathbb{K}v'$, so h_w is injective. The map $V_w \to U_w$, $y \mapsto \pi(y+w)$ is obviously continuous and inverse to h_w since $h_w(\pi(y+w)) = y$ for all $y \in V_w$ and since h_w is injective. So we have proved that $h_w : U_w \to V_w$ is a homeomorphism.

Next observe that all the Hilbert spaces V_w , $w \in S\mathcal{H}$ are pairwise isomorphic since each of them has codimension 1 in \mathcal{H} . After this observation we show that for all $v, w \in S\mathcal{H}$

$$h_w(U_w \cap U_v) = \mathcal{V}_w \setminus (-\operatorname{pr}_{\mathbb{K}^v} w + \mathcal{V}_w \cap \mathcal{V}_v) \quad .$$
(1.1.7)

Assume that $y \in V_w$. The relation $v \notin (-\operatorname{pr}_{\mathbb{K}v} w + V_w \cap V_v)$ then is equivalent to $\operatorname{pr}_{\mathbb{K}v}(y+w) \neq 0$, which on the other hand is equivalent to the existence of some $\lambda \in \mathbb{K}^{\times}$ and $x \in V_v$ such that $y + w = \lambda(x+v)$. Since $h_w^{-1}(y) = \pi(y+w)$, the latter is equivalent to the existence of an $x \in V_v$ such that $h_w^{-1}(y) = \pi(x+v)$. But that is equivalent to $h_w^{-1}(y) \in U_w \cap U_v$. This proves (1.1.7).

The transition map between the chart h_w and the chart h_v is now given by

$$h_v \circ h_w^{-1} : \mathcal{V}_w \setminus \left(-\operatorname{pr}_{\mathbb{K}v} w + \mathcal{V}_w \cap \mathcal{V}_v \right) \to \mathcal{V}_v \setminus \left(-\operatorname{pr}_{\mathbb{K}w} v + \mathcal{V}_w \cap \mathcal{V}_v \right), \, y \mapsto \operatorname{pr}_{\mathcal{V}_v} \frac{y + w}{\langle y + w, v \rangle} \, .$$

But this map is analytic as a composition of analytic maps, hence any two charts are \mathcal{C}^{ω} -compatible. Since $\mathbb{P}\mathcal{H}$ is obviously covered by the open domains U_w , $w \in \mathbb{S}\mathcal{H}$, the projective Hilbert space $\mathbb{P}\mathcal{H}$ becomes an analytic manifold locally modelled on a Hilbert space isomorphic to each of the V_w , $w \in \mathbb{S}\mathcal{H}$. ad (ii) Fix a unit vector $w \in \mathbb{SH}$, let $V_w = (\mathbb{K}w)^{\perp}$ as before and and put

$$\widetilde{\mathbf{V}}_w = \begin{cases} \mathbf{V}_w & \text{if } \mathbb{K} = \mathbb{R} \\ \mathbf{V}_w \oplus \mathbf{i} \mathbb{R} w & \text{if } \mathbb{K} = \mathbb{C} \end{cases},$$

Then $\widetilde{\mathcal{V}}_w$ is the orthogonal complement of the real line $\mathbb{R}w$ with respect to the real inner product $\mathfrak{Re}\langle -, -\rangle$ on \mathfrak{H} . Hence any vector $v \in \mathfrak{H}$ can be uniquely represented in the form $v = v_0 w + \hat{v}$ where $v_0 = \mathfrak{Re}\langle v, w \rangle \in \mathbb{R}$ and $\hat{v} = \operatorname{pr}_{\widetilde{\mathcal{V}}_w}(v) \in \widetilde{\mathcal{V}}_w$. Put $N_w = \mathbb{SH} \setminus \{-w\}$. The stereographic projection

$$g_w: N_w \to \widetilde{\mathcal{V}}_w, \ v \mapsto \frac{2}{1+v_0}\hat{v}$$

then is a chart for $\mathbb{S}\mathcal{H}$ with inverse

$$g_w^-: \widetilde{\mathcal{V}}_w \to N_w, \ z \mapsto \frac{1}{4 + \|z\|^2} \left((4 - \|z\|^2)w + 4z \right)$$

Since $\frac{4-r}{4+r}>-1$ for all $r\geqslant 0$ and

$$\|g_w^-(z)\|^2 = \frac{1}{(4+\|z\|^2)^2} \left((4-\|z\|^2)^2 + (4\|z\|)^2 \right) = 1 \quad \text{for all } z \in \widetilde{\mathcal{V}}_w \ ,$$

the map g_w^- has image in N_w , indeed. Moreover, for $z \in \widetilde{V}_w$,

$$g_w \circ g_w^{-}(z) = \frac{2}{1 + \frac{4 - \|z\|^2}{4 + \|z\|^2}} \frac{4}{4 + \|z\|^2} z = z$$

and for $v \in N_w$ by application of the equality $\|v_0\|^2 + \|\hat{v}\|^2 = 1$,

$$g_w^- \circ g_w(v) = g_w^2 \left(\frac{2}{1+v_0}\hat{v}\right) = \frac{1}{4+\frac{4}{(1+v_0)^2}} \left\|\hat{v}\|^2 \left(4-\frac{4}{(1+v_0)^2}\|\hat{v}\|^2w + \frac{8}{1+v_0}\hat{v}\right) = \frac{1}{(1+v_0)^2+\|\hat{v}\|^2} \left(\left((1+v_0)^2-\|\hat{v}\|^2\right)w + 2(1+v_0)\hat{v}\right) = \frac{1}{2(1+v_0)} \left(2v_0(1+v_0)w + 2(1+v_0)\hat{v}\right) = v_0w + \hat{v} = v .$$

Therefore, g_w are g_w^- mutually inverse as claimed. Observe that for $v \in S\mathcal{H} \setminus \{w\}$ the transition map $g_w \circ g_v^- : \widetilde{V}_v \setminus \{g_v(-w)\} \to \widetilde{V}_w \setminus \{g_w(-v)\}$ is given by

$$z \mapsto g_w \left(\frac{1}{4 + \|z\|^2} \left((4 - \|z\|^2)v + 4z \right) \right) =$$

$$= \frac{2}{1 + \frac{1}{4 + \|z\|^2} \left(\Re \mathfrak{e} \langle (4 - \|z\|^2)v + 4z, w \rangle \right)} \operatorname{pr}_{\widetilde{V}_w} \left(\frac{1}{4 + \|z\|^2} \left((4 - \|z\|^2)v + 4z \right) \right) =$$

$$= \frac{2}{4 + \|z\|^2 + \Re \mathfrak{e} \langle (4 - \|z\|^2)v + 4z, w \rangle} \left((4 - \|z\|^2)v + 4z - \Re \mathfrak{e} \langle (4 - \|z\|^2)v + 4z, w \rangle w \right)$$

which is real analytic. Since the open sets N_w with $w \in SH$ cover the sphere SH it thus becomes a real analytic manifold modelled on a possibly infinite dimensional real Hilbert space. Now consider the composition

$$\widetilde{\mathcal{V}}_w \backslash 2\mathbb{S}\mathcal{V}_w \to \mathcal{V}_w, \quad z \mapsto h_w \circ \pi \circ g_w^-(z) = \operatorname{pr}_{\mathcal{V}_w} \left(\frac{(4 - \|z\|^2)w + 4z}{4 - \|z\|^2 + 4\langle z, w \rangle} \right) = \frac{4\left(z - \langle z, w \rangle w\right)}{4 - \|z\|^2 + 4\langle z, w \rangle} .$$

This is a real analytic map for every $w \in SH$, so $\pi|_{SH}$ is real analytic. Let us show that it is a principal fiber bundle. To this end put $G = \mathbb{Z}/2$ in the real case and $G = S^1$ in the complex case and note that G acts smoothly on SH by scalar multiplication. Since G is abelian, we can write this also as a right action $\cdot : SH \times G \to SH$. By definition of the projective Hilbert space this right action is free and transitive on the fibers of the projection $\pi : SH \to PH$ which therefore are homeomorphic to G. For each $w \in SH$ the map

$$f_w: \mathbb{SH} \setminus \mathcal{V}_w \to U_w \times G \subset \mathbb{PH} \times G, \ v \mapsto \left(\mathbb{K}v, \frac{\langle v, w \rangle}{|\langle v, w \rangle|}\right)$$

now is a bundle trivialization as the following argument shows. By construction, f_w is real analytic with inverse

$$f_w^-: U_w \times G \to \mathbb{SH} \backslash \mathcal{V}_w, \ (\mathbb{K}v, \lambda) \mapsto \lambda \frac{h_w(\mathbb{K}v) + w}{\|h_w(\mathbb{K}v) + w\|}$$

Indeed, f_w is obviuously surjective and

$$f_w^- \circ f_w(v) = \frac{\langle v, w \rangle}{|\langle v, w \rangle|} \frac{h_w(\mathbb{K}v) + w}{\|h_w(\mathbb{K}v) + w\|} = \frac{\langle v, w \rangle}{|\langle v, w \rangle|} \frac{\frac{\langle v, w \rangle}{\langle v, w \rangle}}{\frac{\|v\|}{|\langle v, w \rangle|}} = v \quad \text{for all } v \in \mathbb{SH} \setminus \mathcal{V}_w \ .$$

Observe that $\operatorname{pr}_2 f_w(v \cdot \lambda) = (\operatorname{pr}_2 f_w(v))\lambda$ for all $v \in S\mathcal{H} \setminus V_w$ and $\lambda \in G$, where pr_2 denotes projection onto the second coordinate. Finally note that for $v, w \in S\mathcal{H}$ and $z \in S\mathcal{H} \setminus (V_v \cup V_w)$,

$$f_v \circ f_w^-(\mathbb{K}z,\lambda) = f_v\left(\lambda \frac{h_w(\mathbb{K}z) + w}{\|h_w(\mathbb{K}z) + w\|}\right) = \left(\mathbb{K}z,\lambda \frac{\langle z,v \rangle}{\langle z,w \rangle}\right) = (\mathbb{K}z,\lambda) \cdot \frac{\langle z,v \rangle}{\langle z,w \rangle},$$

where $\cdot : (\mathbb{PH} \times G) \times G \to \mathbb{PH} \times G$ denotes the right action $((\ell, \mu), \lambda)) \mapsto (\ell, \mu) \cdot \lambda = (\ell, \mu\lambda)$. Hence $\pi|_{\mathbb{SH}} : \mathbb{SH} \to \mathbb{PH}$ is a real analytic *G*-principal bundle with local trivializations $f_w, w \in \mathbb{SH}$. \Box

1.1.8 Remark Notice that the chart h_w in the proof of (i) can be written as

$$h_w(\mathbb{K}v) = rac{v}{\langle v, w \rangle} - w \; .$$

This is the same as for the charts of finite dimensional projective space \mathbb{KP}^n . Indeed, we can choose w as a basis element, say e_k , $k = 0, \ldots, n$ and we have a line

$$[v_0:\ldots:v_k:\ldots:v_n] \in \mathbb{KP}^n$$

represented by the vector $v = (v_0, \ldots, v_k, \ldots, v_n)$, where $v_k \neq 0$. Then the standard chart is obtained as follows. First normalize the vector representing the line in the k-th coordinate, i.e. divide by $\langle v, w \rangle$:

$$\left[\frac{v_0}{v_k}:\ldots:1:\ldots:\frac{v_n}{v_k}\right] \;,$$

and then map this to \mathbb{K}^n via dropping the 1 in the k-th coordinate:

$$\left[\frac{v_0}{v_k}:\ldots:1:\ldots:\frac{v_n}{v_k}\right]\mapsto \left(\frac{v_0}{v_k},\ldots,\frac{v_{k-1}}{v_k},\frac{v_{k+1}}{v_k},\ldots,\frac{v_n}{v_k}\right) \ .$$

1.2. Quantum mechanical symmetries

Automorphisms of the projective Hilbert space and Wigner's theorem

1.2.1 Assume that a quantum mechanical system is described by the projective Hilbert space \mathbb{PH} and that two observers \mathcal{O} and \mathcal{O}' observe the system. While observer \mathcal{O} describes the states the system is in by rays $\mathscr{R}, \mathscr{\ell}, \mathscr{\ell}_i, \mathscr{\ell}_i, \ldots \in \mathbb{PH}$, observer \mathcal{O}' describes them by possibly different rays $\mathscr{R}', \mathscr{\ell}', \mathscr{\ell}'_i, \ldots \in \mathbb{PH}$. In other words this means that from the point of physics the rays are not invariant under observer change. Rather does the observer change give rise to a map $A : \mathbb{PH} \to \mathbb{PH}, \ \mathscr{\ell} \mapsto A\mathscr{\ell} = \mathscr{\ell}'$. This map has to be invertible because the observer change is reversible. Even though rays describing the states of the system do change under an observer change, the corresponding transition probabilities remain invariant by the paradigm that the laws of (quantum) physics do not change from one observer to another. Mathematically this can be expressed by

$$\langle \langle A\mathcal{K}, A\mathcal{\ell} \rangle \rangle^2 = \langle \langle \mathcal{K}, \mathcal{\ell} \rangle \rangle^2 \quad \text{for all } \mathcal{K}, \mathcal{\ell} \in \mathbb{PH} .$$

This leads us to the following definition.

1.2.2 Definition Let \mathbb{PH} , \mathbb{PH}_1 and \mathbb{PH}_2 denote projective Hilbert spaces. One then calls a map $A: \mathbb{PH}_1 \to \mathbb{PH}_2$ an *isometry*, if

$$\langle A\mathcal{k}, A\mathcal{\ell} \rangle = \langle \mathcal{k}, \mathcal{\ell} \rangle$$
 for all $\mathcal{k}, \mathcal{\ell} \in \mathbb{PH}_1$.

A bijective isometry $A : \mathbb{PH} \to \mathbb{PH}$ is called an *isometric automorphism*, a Wigner automorphism or just an *automorphism*.

In quantum mechanics, an automorphism of a projective Hilbert space \mathbb{PH} is called a *symmetry* of the quantum mechanical system described by \mathbb{PH} .

1.2.3 Because the composition of isometric maps between projective Hilbert spaces is an isometric map and the identity map on a projective Hilbert space is isometric the projective Hilbert spaces as objects and the isometric maps as morphisms form a category which we call the *Wigner category* denoted it by Wig. The Wigner automorphisms are then the automorphisms of that category.

The automorphisms of a projective Hilbert space \mathbb{PH} form a group denoted by $Aut(\mathbb{PH})$.

1.2.4 From now on in this section let the symbol \mathcal{H} stand for a complex Hilbert space of dimension ≥ 2 . We want to examine what maps on \mathcal{H} induce automorphisms of the corresponding projective Hilbert space.

If $S : \mathcal{H} \to \mathcal{H}$ is a unitary operator that is $S \in \mathsf{GL}(\mathcal{H})$ and $\langle Sv, Sw \rangle = \langle v, w \rangle$ for all $v, w \in \mathcal{H}$, then $\hat{S} : \mathbb{P}\mathcal{H} \to \mathbb{P}\mathcal{H}$, $\mathbb{C}v \mapsto \mathbb{C}Sv$ is well-defined and an automorphism of $\mathbb{P}\mathcal{H}$. But not every automorphism of $\mathbb{P}\mathcal{H}$ is of the form \hat{S} with $S \in U(\mathcal{H})$. Namely let $T : \mathcal{H} \to \mathcal{H}$ be an anti-unitary map that is $T \in \mathsf{GL}(\mathcal{H}, \mathbb{R})$, $T(\lambda v) = \overline{\lambda}Tv$ for all $v \in \mathcal{H}$, $\lambda \in \mathbb{C}$ and $\langle Tv, Tw \rangle = \overline{\langle v, w \rangle} = \langle w, v \rangle$ for all $v, w \in \mathcal{H}$. Then $\hat{T} : \mathbb{P}\mathcal{H} \to \mathbb{P}\mathcal{H}$, $\mathbb{C}v \mapsto \mathbb{C}Tv$ is also well-defined, invertible and preserves transition probabilities. Therefore $\hat{T} \in \mathsf{Aut}(\mathbb{P}\mathcal{H})$. We will later see that \hat{T} is not equal to any of the automorphisms \hat{S} with $S \in U(\mathcal{H})$. Observe also that by the dimension assumption on \mathcal{H} there exists an anti-unitary transformation, for example the real linear map $T : \mathcal{H} \to \mathcal{H}$ which acts on some initially chosen Hilbert basis $(v_j)_{j\in J}$ by $T(v_j) = v_j$ and $T(iv_j) = -iv_j$. One easily checks that the products ST and TS of a unitary operator $S : \mathcal{H} \to \mathcal{H}$ and an anti-unitary operator $T : \mathcal{H} \to \mathcal{H}$ are anti-unitary. If $T_1, T_2 : \mathcal{H} \to \mathcal{H}$ are both anti-unitary, then the product T_1T_2 is unitary. Hence we obtain a new group $AU(\mathcal{H})$ consisting of all unitary and anti-unitary operators on \mathcal{H} . The map

$$\pi: \mathsf{AU}(\mathcal{H}) \to \mathsf{Aut}(\mathbb{P}\mathcal{H}), \ S \mapsto \hat{S}$$

then is a group homomorphism. Its kernel coincides with $U(1) \cong \mathbb{S}^1$. To see this let $\pi(S) = \mathrm{id}_{\mathbb{P}\mathcal{H}}$. Then for every ray ℓ there exists a complex number μ_{ℓ} such that $Sv = \mu_{\mathbb{C}v}v$ for all $v \in \ell$. By unitarity $|\mu_{\ell}| = 1$. Let $v, w \in \mathcal{H}$ be two linearly independent vectors of norm 1. Since

$$\mu_{\mathbb{C}(w-v)}(w-v) = S(w-v) = \mu_{\mathbb{C}w}w - \mu_{\mathbb{C}v}v ,$$

one has $0 = (\mu_{\mathbb{C}(w-v)} - \mu_{\mathbb{C}w})w + (\mu_{\mathbb{C}v} - \mu_{\mathbb{C}(w-v)})v$ which implies $\mu_{\mathbb{C}w} = \mu_{\mathbb{C}(w-v)} = \mu_{\mathbb{C}v}$ by linear independence of v and w. Hence all the $\mu_{\mathbb{C}v}$ coincide and $S = \mu \operatorname{id}_{\mathcal{H}}$ for some complex number $\mu \in U(1) \cong \mathbb{S}^1$. A consequence of this observation is also that the homomorphism $\pi|_{U(\mathcal{H})} : U(\mathcal{H}) \to$ $\operatorname{Aut}(\mathbb{P}\mathcal{H}), S \mapsto \hat{S}$ is not surjective because for every anti-unitary T and unitary S the product TS^{-1} is anti-unitary, hence can not be an element of U(1). We denote the image of $U(\mathcal{H})$ under π by $U(\mathbb{P}\mathcal{H})$ and call its elements the *unitary automorphisms* of $\mathbb{P}\mathcal{H}$.

1.2.5 Theorem (Wigner's theorem, Wigner (1944)) Let \mathcal{H} be a complex Hilbert space of dimension ≥ 2 . Then the sequence of group homomorphisms

$$1 \longrightarrow \mathsf{U}(1) \longrightarrow \mathsf{AU}(\mathcal{H}) \xrightarrow{\pi} \mathsf{Aut}(\mathbb{P}\mathcal{H}) \longrightarrow 1$$

is exact.

1.2.6 Remark Wigner's theorem was first stated in Wigner (1944), but with an incomplete proof. Only several years later complete and independent proofs of Wigner's result were given by Uhlhorn (1962), Lomont & Mendelson (1963), and Bargmann (1964).

Proof. Wigner's theorem is an immediate consequence of the precedinmg considerations and the following more general result. $\hfill \square$

1.2.7 Theorem (Optimal version of Wigner's theorem, Gehér (2014)) Let \mathcal{H} be a complex Hilbert space of dimension ≥ 2 . Then for every isometry $A : \mathbb{PH} \to \mathbb{PH}$ there exists a linear or conjugate-linear isometry $S : \mathcal{H} \to \mathcal{H}$ such that $A = \hat{S}$, where \hat{S} is the isometry on \mathbb{PH} which maps the ray $\mathbb{C}v$ with $v \in \mathcal{H} \setminus \{0\}$ to the ray $\mathbb{C}Sv$.

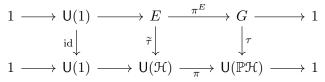
Proof. To prove the claim we will follow the elementary argument by Gehér (2014). \Box

Lifting of projective representations and Bargmann's theorem

1.2.8 Theorem (Bargmann's Theorem) Let \mathcal{H} be a complex Hilbert space and G a connected and simply connected Lie group with $H^2(\mathfrak{g}, \mathbb{R}) = 0$. Then every projective representation $\tau : G \to U(\mathbb{P}\mathcal{H})$ can be lifted to a unitary representation $\sigma : G \to U(H)$ that is $\pi \circ \sigma = \tau$, where $\pi : U(\mathcal{H}) \to U(\mathbb{P}\mathcal{H})$ is the canonical projection.

1.2.9 Remark The lifting theorem was proved first in Bargmann (1954). The short proof we present here goes back to Simms (1971). We closely follow his argument.

Proof of the theorem. Let E be the fibered product of π and τ with the canonical homomorphisms $\tilde{\tau}: E \to U(\mathcal{H})$ and $\pi^E: E \to G$. For the resulting commutative diagram of groups with two exact rows



we want to construct a section $s: G \to E$ of $\pi^E : E \to G$ which is a splitting meaning that s is a group homomorphism and $\pi^E \circ s = id_G$. With the construction of such an s we are done because then the unitary representation $\tilde{\tau} \circ s$ is a lifting of the projective representation $\tau : G \to U(\mathbb{PH})$.

Observe that E is a Lie group by Kuranishi's theorem, see (Montgomery & Zippin, 1955, §4.3), since E is central extension of a Lie group, hence locally compact, and there exist local continuous sections $\sigma: U \to E$ that is $U \subset G$ is open and $\pi^E \circ \sigma = \mathrm{id}_U$.

The short exact sequence of Lie groups

$$1 \longrightarrow \mathsf{U}(1) \longrightarrow E \xrightarrow{\pi^E} G \longrightarrow 1$$

induces a short exact sequence of Lie algebras

$$0 \longrightarrow \mathbb{R} \longrightarrow \mathfrak{e} \xrightarrow{T\pi^E} \mathfrak{g} \longrightarrow 0 , \qquad (1.2.1)$$

where \mathfrak{e} is the Lie algebra of E and \mathfrak{g} the one of G. Observe that $T\pi^E$ is surjective with kernel \mathbb{R} being in the center of \mathfrak{e} . Choose a linear map $\lambda : \mathfrak{g} \to \mathfrak{e}$ such that $\pi^E \circ \lambda = \mathrm{id}_{\mathfrak{g}}$. Put $\Theta(x, y) = [\lambda(x), \lambda(y)] - \lambda([x, y])$ for all $x, y \in \mathfrak{g}$. Then

$$T\pi^E \circ \Theta(x,y) = [T\pi^E \circ \lambda(x), T\pi^E \circ \lambda(y)] - T\pi^E \circ \lambda([x,y]) = [x,y] - [x,y] = 0$$

Hence $\Theta(x, y)$ is in the kernel of $T\pi^E$ which means that Θ is a map $\mathfrak{g} \times \mathfrak{g} \to \mathbb{R}$. By definition, $\Theta : \mathfrak{g} \times \mathfrak{g} \to \mathbb{R}$ is skew symmetric. Let us show that it satisfies the Jacobi identity. Compute, using the Jacobi identity for the Lie algebra bracket and the fact that Θ has image in the center of \mathfrak{e} ,

$$\begin{split} \Theta([x,y]),z) + \Theta([y,z],x) + \Theta([z,x],y) &= \\ &= [\lambda([x,y]),\lambda(z)] + [\lambda([y,z]),\lambda(x)] + [\lambda([z,x]),\lambda(y)] - \\ &- \lambda([[x,y],z]) - \lambda([[y,z],x]) - \lambda(([z,x],y]) = \\ &= [[\lambda(x),\lambda(y)],\lambda(z)] + [[\lambda(y),\lambda(z)],\lambda(x)] + [[\lambda(z),\lambda(x)],\lambda(y)] - \\ &- [\Theta([x,y]),\lambda(z)] - [\Theta([y,z]),\lambda(x)] - [\Theta([z,x]),\lambda(y)] = 0 \;. \end{split}$$

Therefore, Θ is a Lie algebra 2-cocycle. By $H^2(\mathfrak{g}, \mathbb{R}) = 0$, there exists a linear $\theta : \mathfrak{g} \to \mathbb{R}$ such that $\Theta(x, y) = \theta([x, y])$ for all $x, y \in \mathfrak{g}$. Put $\mu(x) = \lambda(x) + \theta(x)$. Then, since θ has values in the center of \mathfrak{e} ,

$$\begin{split} \left[\mu(x),\mu(y)\right] &= \left[\lambda(x) + \theta(x),\lambda(y) + \theta(y)\right] = \left[\lambda(x),\lambda(y)\right] = \\ &= \Theta(x,y) + \lambda([x,y]) = \theta([x,y]) + \lambda([x,y]) = \mu([x,y]) \;. \end{split}$$

Hence $\mu : \mathfrak{g} \to \mathfrak{e}$ is a Lie-Algebra homomorphism and fulfills

$$T\pi^E \circ \mu(x) = T\pi^E(\lambda(x) + \theta(x)) = T\pi^E(\lambda(x)) = x \quad \text{for all } x \in \mathfrak{g} \ .$$

So μ is also a section of $T\pi^E$ which shows that the short exact sequence of Lie algebras (1.2.1) is split.

By $\pi_1(G) = 1$, the Lie algebra homomorphism $\mu : \mathfrak{g} \to \mathfrak{e}$ has a lifting to a group homomorphism $s : G \to E$ such that $\pi^E \circ s = \mathrm{id}_{\mathfrak{g}}$. The proof is finished.

II.2. Deformation quantization

2.1. Fedosov's construction of star products

The various Weyl algebras of a Poisson vector space

2.1.1 Definition By a *Poisson vector space* over the field \mathbb{K} of real or complex numbers one understands a pair (V, Π) where V is a finite dimensional vector space over \mathbb{K} and $\Pi \in \Lambda^2 V$ is an antisymmetric bivector.

Given two Poisson vector spaces (V, Π) and (W, Ξ) , a linear map $f : V \to W$ is called a *morphism* of Poisson vector spaces if $f_*\Pi := (f \otimes f)\Pi = \Xi$.

Poisson vector spaces together with their morphisms obviously form a category which we denote by $PVec_{\mathbb{K}}$.

2.1.2 Example Let $V = \mathbb{R}^{2n}$ or $V = \mathbb{R}^{2n+1}$. Then V together with the bivector $\Pi_{can} = \sum_{k=1}^{n} \frac{\partial}{\partial x_{k+n}} \wedge \frac{\partial}{\partial x_k}$ is a Poisson vector space. One calls Π_{can} the canonical (constant) Poisson structure on V.

2.1.3 Let $rk \Pi$ be the rank of Π that is the dimension of the image of the musical map

$$\Pi^{\sharp}: V^* \to V, \ \alpha \mapsto \alpha \,\lrcorner\, \Pi \ ,$$

where

$$\alpha \sqcup : \Lambda^k V \to \Lambda^{k-1} V, \sum_{i=1}^N v_{i,1} \land \ldots \land v_{i,k} \mapsto \sum_{i=1}^N \sum_{j=1}^k (-1)^{j+1} \langle \alpha, v_{i,j} \rangle \land v_{i,1} \land \ldots \land \widehat{v_{i,j}} \land \ldots \land v_{i,k}$$

denotes the interior product of a 1-form with an alternating k-vector. Then $\operatorname{rk}\Pi$ is even dimensional, and (V,Π) isomorphic as a Poisson vector space to the product of $(\mathbb{R}^{\operatorname{rk}\Pi},\Pi_{\operatorname{can}})$ with $(\mathbb{R}^{\dim V-\operatorname{rk}\Pi},0)$.

2.1.4 Remark The category $PVec_{\mathbb{K}}$ is dual to the category $PSVec_{\mathbb{K}}$ of presymplectic vector spaces over \mathbb{K} that is the category of all finite dimensional \mathbb{K} -vector spaces W together with a (constant) 2-form $\omega \in \Lambda^2 W^*$.

A contravariant isomorphism between these two categories is given by the dualization functor *: $\mathsf{PVec}_{\mathbb{K}} \to \mathsf{PSVec}_{\mathbb{K}}$ which maps $V \mapsto V^*$ and the bivector Π on V to the 2-form $\omega : V^* \times V^* \to \mathbb{K}$, $(\alpha, \beta) \mapsto \beta \lrcorner (\alpha \lrcorner \Pi)$. Its inverse is again given by dualization. **2.1.5** The bivector Π of a Poisson vector space (V, Π) turns V into a Poisson manifold with bracket $\{-, -\} : \mathcal{C}^{\infty}(V) \times \mathcal{C}^{\infty}(V) \to \mathcal{C}^{\infty}(V)$ given by

$$\{f,g\} = dg \,\lrcorner\, (df \,\lrcorner\, \Pi) \quad \text{for } f,g \in \mathcal{C}^{\infty}(V) \;.$$

By construction, $\{-,-\}$ is antilinear and a derivation in each component. Since for all linear functions $\lambda, \mu : V \to \mathbb{K}$ the Poisson bracket $\{\lambda, \mu\}$ is constant, the Poisson bracket $\{\{\lambda, \mu\}, \nu\}$ of three linear functions vanishes, hence the Jacobi identity holds for linear and affine functions. This implies that the Jacobi identity is satisfied for all smooth functions, hence $\{-,-\}$ is a Poisson bracket on V indeed. We call it the *constant Poisson structure* associated to Π .

2.1.6 Definition The Weyl algebra of a Poisson vector space (V, Π) is defined by

$$\mathsf{A}(V,\Pi) = \mathsf{T}^{\bullet}V^* / (\alpha \otimes \beta - \beta \otimes \alpha - \beta \lrcorner (\alpha \lrcorner \Pi) \mid \alpha, \beta \in V^*) ,$$

where (X) stands for the ideal generated by $X \subset \mathsf{T}^{\bullet}V^*$.

2.1.7 Remarks (a) To a presymplectic vector space (W, ω) one associates the Weyl algebra

$$\mathsf{A}(W,\omega) = \mathsf{T}^{\bullet}W/(v \otimes w - w \otimes v - w \llcorner (v \llcorner \omega) \mid v, w \in W) ,$$

where $\[\]$ denotes the interior product of a vector with a k-form. If (W, ω) is the dual of a Poisson vector space (V, Π) , then the two Weyl algebras $A(V, \Pi)$ and $A(W, \omega)$ coincide by definition. We will silently make use of this fact in the following considerations.

(b) Let K be a field of characteristic 0 and K[x₁,...,x_n] the polynomial ring over K in n (commuting) indetereminates. The n-th Weyl algebra A_n(K) over K is then defined as the subalgebra of the endomorphism ring End_K(K[x₁,...,x_n]) generated by the elements

$$\hat{x}_k : \mathbb{K}[x_1, \dots, x_n] \to \mathbb{K}[x_1, \dots, x_n], \ p \mapsto x_k \cdot p$$

and

$$\partial_k : \mathbb{K}[x_1, \dots, x_n] \to \mathbb{K}[x_1, \dots, x_n], \ p \mapsto \frac{\partial p}{\partial x_k}$$

where k runs through $1, \ldots, n$. The commutation relations for these operators are, using the Kronecker delta,

$$[\hat{x}_k, \hat{x}_l] = 0, \quad [\partial_k, \partial_l] = 0, \quad [\partial_k, \hat{x}_l] = \delta_{k,l} . \tag{2.1.1}$$

Recall that $A_n(\mathbb{K})$ coincides with the ring of differential operators on $\mathbb{K}[x_1, \ldots, x_n]$ in the sense of Grothendieck. For a proof see Coutinho (1995).

(c) Let ω be a the canonical symplectic form on \mathbb{R}^{2n} . The Weyl algebra $A(\mathbb{R}^{2n}, \omega)$ then coincides naturally with the algebra of differential operators on \mathbb{R}^n with polynomial coefficients. To see this denote the canonical basis of \mathbb{R}^{2n} by $(Q_1, \ldots, Q_n, P_1, \ldots, P_n)$ and the corresponding coordinate functions by $(q_1, \ldots, q_n, p_1, \ldots, p_n)$. The commutators of these basis elements in the Weyl algebra are

$$[Q_k, Q_l] = 0, \quad [P_k, P_l] = 0, \quad [P_k, Q_l] = \delta_{k,l} .$$
(2.1.2)

Therefore, any element of $\mathsf{A}(\mathbb{R}^{2n},\omega)$

Next consider the symmetric (covariant) tensor algebra $S^{\bullet}V^*$ over V. Recall that it is defined as the algebra with underlying vector space

$$\mathsf{S}^{\bullet}V^* = \bigoplus_{k \in \mathbb{N}} \mathsf{S}^k V^* , \qquad (2.1.3)$$

where $S^k V^* \subset \bigotimes^k V^*$ denotes the space of all symmetric (covariant) k-tensors in V. An element $t \in S^k V^*$ is called *homogenous* of symmetric degree deg_s t = k. It can be written in the form

$$t = \sum_{i \in I} t_{i,1} \otimes \ldots \otimes t_{i,k}$$

where I is a finite index set, and $t_{i,1}, \ldots, t_{i,k}$ are elements of the dual V^* .

The bundle of formal Weyl algebras

Let M be a smooth manifold. Recall the notion of the symmetric (covariant) tensor algebra bundle $S^{\bullet}T^*M$ over M. It is defined by

$$\mathsf{S}^{\bullet}T^*M = \bigoplus_{k \in \mathbb{N}} \mathsf{S}^k T^*M , \qquad (2.1.4)$$

where $S^kT^*M = \bigcup_{p \in M} S^kT^*_pM \subset \bigotimes^k T^*M$ is the bundle of all symmetric (covariant) *k*-tensors. Note that we have a canonical (fiberwise) isomorphism $S^kT^*M \cong (S^kTM)^*$ which leads to the canonical identifications

$$\mathbf{S}^{\bullet}M = \bigoplus_{k \in \mathbb{N}} \mathbf{S}_k T^* M \cong \bigoplus_{k \in \mathbb{N}} \mathbf{S}^k T M = \mathbf{S}^{\bullet} T M$$

An element $t \in S^k T^*M$ is called *homogenous* of *symmetric degree* $\deg_s t = k$. It can be written in the form

$$t = \sum_{i \in I} t_{1,i} \otimes \ldots \otimes t_{k,i} ,$$

where I is a finite index set, and $t_{1,i}, \ldots, t_{k,i}$ are elements of the cotangent bundle T^*M having the same footpoint as t. Every element of the symmetric tensor algebra bundle S[•]M can be expanded as a finite sum of homogeneous symmetric tensors.

The (fiberwise) symmetric product $\vee : SM \times_M SM \to SM$ is constructed by defining it, for each $p \in M$, first on homogeneous elements $t = \sum_{i \in I} t_{1,i} \otimes ... \otimes t_{k,i} \in S_p^k M$ and $s = \sum_{j \in J} s_{k+1,j} \otimes ... \otimes s_{k+l,j} \in S_p^l M$ by

$$v(t,s) = t \lor s = \frac{1}{k! \, l!} \sum_{\sigma \in S_{k+l}} \sum_{i \in I, j \in J} v_{\sigma(1), ij} \otimes \dots \otimes v_{\sigma(k+l), ij} , \text{ where}$$

$$v_{m,ij} = \begin{cases} t_{m,i} & \text{if } 1 \leqslant m \leqslant k , \\ s_{m,j} & \text{if } k < m \leqslant k+l , \end{cases}$$

$$(2.1.5)$$

and then extending it linearly in each component to the whole fiber $S_p^{\bullet}M \times S_p^{\bullet}M$. Using the canonical symmetrization operator

$$\mathsf{S}:\mathsf{T}^{\bullet}M=\mathsf{T}^{\bullet}TM\to\mathsf{S}M,\ t=\sum_{i\in I}t_{1,i}\otimes\ldots\otimes t_{k,i}\mapsto\sum_{\sigma\in S_k}\sum_{i\in I}t_{\sigma(1),i}\otimes\ldots\otimes t_{\sigma(k),i}$$

we can also write

$$t \lor s = \binom{k+l}{k} \mathsf{S}(t \otimes s). \tag{2.1.6}$$

Together with the symmetric product SM now becomes a graded algebra. Note that it is canonically isomorphic to the algebra $C^{\infty}_{pol}(TM)$ of smooth functions on TM which are polynomial in the fibers of TM.

Let us define an action of an antisymmetric bivector field $B = \sum_{\iota} B^1_{\iota} \otimes B^2_{\iota} \in \Omega^2 M$ on $SM \otimes SM$ by

$$SM \otimes SM \ni t \otimes s \mapsto B(t \otimes s) = \sum_{\iota} B^{1}_{\iota} \downarrow t \otimes B^{2}_{\iota} \lrcorner s \in SM \otimes SM.$$
(2.1.7)

Under the isomorphism $SM \to C^{\infty}_{pol}(TM)$ the bivector field B acts as a bidifferential operator, i.e. we have for $f, g \in C^{\infty}_{pol}(TM)$, $v, w \in T_xM$ and $x \in M$

$$B(f \otimes g)(v,w) = \sum_{\iota} B_{\iota}^{1} f(v) \otimes B_{\iota}^{2} g(w) = \sum_{\iota} \left. \frac{d}{dt} f(v+tB_{\iota}^{1}) \right|_{t=0} \left. \frac{d}{ds} g(w+sB_{\iota}^{2}) \right|_{s=0}.$$
 (2.1.8)

With these preparations in mind we are now able to define Fedosov's notion of the bundle of formal Weyl algebras.

2.1.8 Definition Let (M, ω) be a symplectic manifold of dimension 2n and Π the corresponding Poisson bivector. The *formal Weyl algebra* AM of M is then defined as the space $SM[[\hbar]]$ of formal power series with coefficients in SM together with the *Moyal product* \circ given by

$$f \circ g = \sum_{k \in \mathbb{N}} \frac{1}{k!} \left(\frac{i\hbar}{2} \right)^k \vee \left(\Pi^k \left(f \otimes g \right) \right) = \vee \left(\exp \left(\frac{i\hbar}{2} \Pi \right) f \otimes g \right).$$
(2.1.9)

Note that in this definition all operations on SM were naturally extended to $SM[[\hbar]]$.

On the Weyl algebra bundle AM we introduce the Fedosov filtration

$$\mathsf{A}M = \mathsf{A}^0 M \subset \mathsf{A}^1 M \subset \mathsf{A}^2 M \subset \dots \subset \mathsf{A}^k M \subset \dots$$
(2.1.10)

by defining

$$\mathsf{A}^{k}M = \left\{ t = \sum_{l,r \in \mathbb{N}} t_{rl} \,\hbar^{l} \in \mathsf{S}M[[\hbar]] : t_{rl} \in \mathsf{S}^{r}M \& t_{rl} = 0 \text{ for } r + 2l < k \right\}.$$
 (2.1.11)

The topology generated by this filtration is called the *F*-topology. Furthermore we define the *F*-degree $\deg_F t$ of an element $t \in AM$ as the supremum of all $k \in \mathbb{N}$ with $t \in A^k M$.

By definition deg_E $0 = \infty$, deg_E $\hbar = 2$ and deg_E $\lambda = m$ for any covariant *m*-tensor field λ .

We have to show that \circ is a well-defined product on $SM[[\hbar]]$ and that the A^kM define a filtration on the algebra AM indeed. It suffices to show that \circ is associative and that $A^kM \circ A^lM \subset A^{k+l}M$

holds for all $k, l \in \mathbb{N}$. Associativity of \circ follows from the following chain of equalities:

$$(f \circ g) \circ h = \sum_{k \in \mathbb{N}} \sum_{l \in \mathbb{N}} \left(\frac{i\hbar}{2} \right)^{k+l} \vee \Pi^k \left(\vee \Pi^l \left(f \otimes g \right) \otimes h \right)$$

$$= \sum_{r \in \mathbb{N}} \sum_{k+l=r} \left(\frac{i\hbar}{2} \right)^r \vee \Pi^k \left(\vee \Pi^l \left(f \otimes g \right) \otimes h \right)$$

$$= \sum_{r \in \mathbb{N}} \sum_{k+l+m=r} \left(\frac{i\hbar}{2} \right)^r \vee \left(\Pi^k_{13} \Pi^l_{23} \Pi^m_{12} \left(f \otimes g \otimes h \right) \right)$$

$$= \sum_{r \in \mathbb{N}} \sum_{k+l+m=r} \left(\frac{i\hbar}{2} \right)^r \vee \left(\Pi^k_{13} \Pi^m_{12} \Pi^l_{23} \left(f \otimes g \otimes h \right) \right)$$

$$= \sum_{r \in \mathbb{N}} \sum_{k+l=r} \left(\frac{i\hbar}{2} \right)^r \vee \Pi^l \left(f \otimes \vee \Pi^k \left(g \otimes h \right) \right)$$

$$= f \circ (g \circ h).$$

$$(2.1.12)$$

Here we have denoted by $\Pi_{\iota\kappa} (a \otimes b \otimes c)$ the natural action of Π on the ι, κ -factors of $a \otimes b \otimes c$ and have used the Jacobi-identity for the Poisson bivector Π . The second claim follows immediately from Eq. (2.1.9) and the definition of $\mathsf{A}^k M$.

to do:By definition SM is a graded $\mathcal{C}^{\infty}(M)$ -module.

Besides AM we will consider in the following differential forms with values in AM, i.e. we will consider the space $\Omega AM := AM \otimes \Omega M \cong (SM \otimes \Omega M)[[\hbar]] = (SM \otimes \Omega M)^{\mathbb{N}}$. By \circ and the exterior product on ΩM this vector space carries a multiplicative structure which also will be denoted by \circ . A second multiplicative structure, which we denote by \cdot , comes from the symmetric product on SM and the exterior product on ΩM . The filtration on AM induces one on ΩAM by

$$\Omega \mathsf{A} M \subset \mathsf{A}^1 M \otimes \Omega M \subset \dots \subset \mathsf{A}^k M \otimes \Omega M \subset \dots;$$
(2.1.13)

thus making $(\Omega AM, \circ)$ into a filtered algebra. Additionally ΩAM posseses a graduation coming from ΩM :

$$\Omega \mathsf{A}M = \bigoplus_{1 \leqslant q \leqslant 2n} \mathsf{A}M \otimes \Omega^q M. \tag{2.1.14}$$

The corresponding degree function $\Omega AM \to \mathbb{R}^{2n}$ will be denoted by \deg_a , the *antisymmetric degree*. Together with the symmetric degree ΩAM now becomes a bigraded vector space. Therefore we have for any element $a \in \Omega AM$ a decomposition

$$a = \sum_{pq} a_{pq}, \qquad (2.1.15)$$

where a_{pq} is the unique homogeneous component of a with symmetric degree p and antisymmetric degree q or in other words with bidegree (p,q). With respect to the product \cdot , but not \circ , ΩAM becomes a bigraded algebra. Nevertheless $(\Omega AM, \circ)$ is a graded algebra with respect to the antisymmetric degree.

Next we introduce the \circ -supercommutator [-,-] on ΩAM as the unique bilinear map such that for two elements $a, b \in \Omega AM$ being homogeneous with respect to the antisymmetric degree the equation

$$[a,b] = a \circ b - (-1)^{\deg_a a \cdot \deg_a b} b \circ a$$
(2.1.16)

holds. The supercommutator induces for every $a \in \Omega AM$ an adjoint map

ad
$$(a)$$
: $\Omega AM \to \Omega AM, b \mapsto [a, b].$ (2.1.17)

Moreover $\frac{1}{\hbar} \operatorname{ad}(a)$ is a well-defined map on ΩAM and comprises a superderivation of ΩAM . The symplectic form $\omega = \sum_{ij} \omega_{ij} dx_i \otimes dx_j$ can be interpreted as an element of $AM \otimes \Omega^1 M$. Thus it gives rise to the inner superderivation

$$\delta = -\frac{i}{\hbar} \operatorname{ad} \left(\omega \right) \tag{2.1.18}$$

of ΩAM . Let us denote for any smooth vector field $V \in C^{\infty}(TM)$ and every element $f \otimes \alpha \in AM \otimes \Omega M$ the insertion $(V \sqcup f) \otimes \alpha$ (resp. $f \otimes (V \sqcup \alpha)$) of V in the symmetric (resp. antisymmetric) part of $f \otimes \alpha$ by $V \sqcup_{s} (f \otimes \alpha)$ (resp. $V \sqcup_{a} (f \otimes \alpha)$). With this notation we get the following expansion of δ in local coordinates:

$$\delta(a) = -\frac{i}{\hbar} \left(\omega \circ a - (-1)^{k} a \circ \omega \right)$$

$$= -\frac{i}{\hbar} \underbrace{\left(\omega \circ a - (-1)^{k} a \circ \omega \right)}_{=0} + \frac{1}{2} \sum_{kl} \prod_{kl} \omega \left(\frac{\partial}{\partial x_{k}}, - \right) \cdot \left(\frac{\partial}{\partial x_{l}} \right) \sqcup a - (-1)^{k} \prod_{kl} \left(\frac{\partial}{\partial x_{k}} \sqcup a \right) \cdot \omega \left(\frac{\partial}{\partial x_{l}}, - \right)$$

$$= \sum_{l} \left(1 \otimes dx_{l} \right) \cdot \left(\frac{\partial}{\partial x_{l}} \sqcup a \right).$$
(2.1.19)

Here we have used the local expansion

$$\Pi = \sum_{kl} \Pi_{kl} \frac{\partial}{\partial x_k} \otimes \frac{\partial}{\partial x_l}$$
(2.1.20)

and the fact that

$$\sum_{k} \Pi_{kl} \,\omega\left(\frac{\partial}{\partial x_k}, -\right) = dx_l. \tag{2.1.21}$$

Analogously we can define a second operator δ^* on ΩAM by setting locally

$$\delta^*(a) = \sum_l \left(dx_l \otimes 1 \right) \cdot \left(\frac{\partial}{\partial x_l} \, \llcorner_{\mathsf{a}} a \right). \tag{2.1.22}$$

 $\delta^*(a)$ is well-defined, as it can be written in the form

$$\delta^*(a) = e(\vee \otimes \llcorner)a,\tag{2.1.23}$$

where $e \in \mathcal{C}^{\infty}(TM)$ is the *Euler tensor field* which locally is given by $e = \sum_{l} dx_{l} \otimes \frac{\partial}{\partial x_{l}}$. Note that δ^{*} is not a superderivation of ΩAM .

2.1.9 Proposition The operators δ and δ^* are homogeneous of symmetric degree -1 (resp. 1) and antisymmetric degree 1 (resp. -1). Moreover they fulfill the following two relations:

$$\delta^2 = (\delta^*)^2 = 0, \tag{2.1.24}$$

$$(\delta \,\delta^* + \delta^* \delta)(f \otimes \alpha) = (p+q)(f \otimes \alpha), \tag{2.1.25}$$

where $f \in AM$ is homogeneous of symmetric degree p and $\alpha \in \Omega^q M$.

Proof. The first property follows from the local expressions for δ and δ^* :

$$\delta^{2}(f \otimes \alpha) = \sum_{kl} \left(\frac{\partial}{\partial x_{k}} \vee \frac{\partial}{\partial x_{l}} \right) f \otimes dx_{k} \wedge dx_{l} \wedge \alpha = 0, \qquad (2.1.26)$$

$$\delta^{*2}(f \otimes \alpha) = \sum_{kl} \left(dx_k \vee dx_l \right) \otimes \left(\frac{\partial}{\partial x_k} \wedge \frac{\partial}{\partial x_l} \right) \, \alpha = 0, \tag{2.1.27}$$

as both sums are symmetric and antisymetric with respect to the indices k, l. The second property is also a direct consequence of the local expressions for δ and δ^* .

Denote by δ^- : $\Omega AM \to \Omega AM$ the operator

$$\Omega \mathsf{A}M \ni a = \sum_{pq} a_{pq} \mapsto \delta^{-}(a) = \sum_{p+q>0} \frac{1}{p+q} \,\delta^* a_{pq} \in \Omega \mathsf{A}M.$$
(2.1.28)

Then the above proposition entails a kind of Hodge-De Rham decomposition in ΩAM , namely the relation

$$a = \delta \,\delta^{-}(a) + \delta^{-} \,\delta(a) + a_{00}. \tag{2.1.29}$$

for every $a \in \Omega AM$.

In the following the notion of the \circ -center $Z(_{)} \circ M$ of ΩAM will be very useful. It is defined as the kernel of the family $(ad(a))_{a \in \Omega AM}$ and obviously fulfills the equation

$$Z(_{0} \circ M = \mathsf{S}^{0} M \otimes \Omega M = \{a \in \Omega \mathsf{A} M : \deg_{\mathsf{s}} a = 0\}.$$
(2.1.30)

There are two canonical projections from ΩAM in $Z(_0 \circ M$, namely

$$\pi_{00}: \ \Omega \mathsf{A}M \to \Omega \mathsf{A}M, \quad a = \sum_{pq} a_{pq} \mapsto a_{00} \tag{2.1.31}$$

and

$$\pi_0: \ \Omega \mathsf{A}M \to \Omega \mathsf{A}M, \quad a = \sum_{pq} a_{pq} \mapsto \sum_q a_{0q}.$$
(2.1.32)

Connections on the formal Weyl algebra

We now want to give ΩAM some more differential geometric structure. To achieve this let us choose a symplectic connection ∇ on M, i.e. a connection ∇ fulfilling $\nabla \omega = 0$. Then ∇ gives rise to a connection ∇ on ΩAM by defining

$$\nabla(f \otimes \alpha) = \nabla f \cdot \alpha + f \otimes d\alpha \tag{2.1.33}$$

for $f \in AM$ and $\alpha \in \Omega M$. Hereby we naturally regard ∇f as an element of $\Omega^1 AM$. As ∇ is supposed to be torsionfree, we have $d\alpha = \nabla \alpha$, so $\nabla : \Omega AM \to \Omega AM$ is a connection on ΩAM indeed, i.e. it fulfills

$$\nabla(\varphi a) = (1 \otimes d\varphi) \cdot a + \varphi \, Da \tag{2.1.34}$$

for every $a \in \Omega AM$ and $\varphi \in C^{\infty}(M)$. Moreover, ∇ is a homogeneous superderivation of $(\Omega AM, \cdot)$ with bidegree (0, 1), as the equation

$$\nabla \left((f \otimes \alpha) \cdot (g \otimes \beta) \right) = \nabla \left((f \vee g) \otimes (\alpha \wedge \beta) \right)$$

= $\left(\nabla f \cdot g + f \cdot \nabla g \right) \cdot (\alpha \wedge \beta) + (f \vee g) \cdot \left(d\alpha \wedge \beta + (-1)^{\deg_{a} \alpha} \alpha \wedge d\beta \right)$
= $\left(\nabla f \cdot \alpha + f \otimes d\alpha \right) \cdot (g \otimes \alpha) + (-1)^{\deg_{a} \alpha} (f \otimes \alpha) \cdot (\nabla g \cdot \beta + g \otimes d\beta)$
= $\left(\nabla (f \otimes \alpha) \right) \cdot (g \otimes \beta) + (-1)^{\deg_{a} \alpha} (f \otimes \alpha) \cdot (\nabla (g \otimes \beta))$
(2.1.35)

holds for homogeneous $f \otimes \alpha, g \otimes \beta \in \Omega AM$. With respect to *, the connection ∇ is a homogeneous superderivation of antisymmetric degree 1 as well. To prove this first recall that $\nabla \Pi = 0$, hence

$$\nabla(f * g) = (\nabla f) * g + f * (\nabla g). \tag{2.1.36}$$

But then

$$\nabla (f \otimes \alpha) * (g \otimes \beta)) = \nabla (f * g) \cdot (\alpha \wedge \beta) + (f * g) \otimes d(\alpha \wedge \beta)$$

= $((\nabla f) * g + f * (\nabla g)) \cdot (\alpha \wedge \beta) + (f * g) \otimes (d\alpha \wedge \beta + (-1)^{\deg_{a}\alpha} \alpha \wedge d\beta)$
= $(\nabla f \cdot \alpha + f \otimes d\alpha) * (g \otimes \alpha) + (-1)^{\deg_{a}\alpha} (f \otimes \alpha) * (\nabla g \cdot \beta + g \otimes d\beta)$
= $(\nabla (f \otimes \alpha)) * (g \otimes \beta) + (-1)^{\deg_{a}\alpha} (f \otimes \alpha) * (\nabla (g \otimes \beta))$
(2.1.37)

which gives the claim.

2.1.10 Proposition The *-superderivation ∇ fulfills the following relations:

$$[\nabla, \delta] = \nabla \delta + \delta \nabla = 0, \qquad (2.1.38)$$

$$[\nabla, \nabla] = 2\nabla^2 = 2\frac{i}{\hbar} \text{ ad } (\tilde{R}), \qquad (2.1.39)$$

where $\tilde{R} \in ST^*M \otimes \Omega^2 M$ is the contraction $\tilde{R} = \omega \llcorner R$ of the curvature tensor R of ∇ . Furthermore the contracted curvature \tilde{R} satisfies the relation

$$\nabla R = \delta R = 0. \tag{2.1.40}$$

Proof. By $\nabla \omega = 0$ we have

$$[\nabla, \delta] = -\frac{i}{\hbar} [\nabla, \mathrm{ad} (\omega)] = -\frac{i}{\hbar} (\nabla \omega) = 0.$$
(2.1.41)

As ∇ has antisymmetric degree 1, the supercommutator of ∇ with itself is equal to $2\nabla^2$. But now we have in local coordinates

$$2\nabla^{2}(f \otimes \alpha) = 2\nabla(\nabla f \cdot \alpha + f \otimes d\alpha) = 2\nabla^{2}f \cdot \alpha$$

$$= \sum_{rs} (\nabla_{\partial_{r}} \nabla_{\partial_{s}} - \nabla_{\partial_{s}} \nabla_{\partial_{r}})f \otimes dx_{r} \wedge dx_{s} \wedge \alpha$$

$$= -\sum_{klrs} R^{k}{}_{lrs} dx_{l} \vee (\partial_{k} \llcorner f) \otimes dx_{r} \wedge dx_{s} \wedge \alpha$$

(2.1.42)

and

$$2\frac{i}{\hbar} \operatorname{ad} \left(\tilde{R}\right) \left(f \otimes \alpha\right) = 2\frac{i}{\hbar} \left(\tilde{R} * \left(f \otimes \alpha\right) - \left(f \otimes \alpha\right) * \tilde{R}\right) = \\ = -\frac{1}{2} \left(\sum_{klmrs} \Pi_{mk} \tilde{R}_{mlrs} dx_l \lor \left(\partial_k \llcorner f\right) \otimes dx_r \land dx_s \land \alpha - \\ -\sum_{klmrs} \Pi_{mk} \tilde{R}_{klrs} \left(\partial_m \llcorner f\right) \lor dx_l \otimes \alpha \land dx_r \land dx_s\right) \\ = -\sum_{klrs} R^k_{lrs} dx_l \lor \left(\partial_k \llcorner f\right) \otimes dx_r \land dx_s \land \alpha,$$

$$(2.1.43)$$

which gives the second equation. The relation $\nabla \tilde{R} = 0$ is nothing else but the first Bianchi identity for the connection ∇ . Last we have

$$\delta \tilde{R} = \frac{1}{2} \sum_{klrs} R_{klrs} \, dx_l \otimes dx_k \wedge dx_r \wedge dx_s = 0, \qquad (2.1.44)$$

as R_{klrs} is cyclic with respect to the indices (l, r, s).

Besides
$$\nabla$$
 we will also consider more general connections on ΩAM , in particular connections D :
 $\Omega AM \rightarrow \Omega AM$ of the form

$$D = \nabla + \frac{i}{\hbar} \operatorname{ad} (\gamma), \qquad (2.1.45)$$

where γ is an element of $\Omega^1 AM$, uniquely determined by D up to a central one-form. We call such a D a *Weyl connection* and attach to it a now unique one-form γ_D fulfilling Eq. (2.1.45) and the normalization condition

$$\pi_0(\gamma_D) = 0. \tag{2.1.46}$$

The two-form

$$\Omega = \tilde{R} + \nabla \gamma_D + \frac{i}{\hbar} \gamma_D * \gamma_D$$
(2.1.47)

will then be called the *Weyl curvature* of D. Furthermore a Weyl connection D will be called *abelian*, if its Weyl curvature is a central form or, using the following proposition, if

$$D^2 = \frac{i}{\hbar} \operatorname{ad} (\tilde{\Omega}) = 0.$$
 (2.1.48)

2.1.11 Proposition Let D be a Weyl connection on ΩAM and Ω its Weyl curvature. Then Ω fulfills the Bianchi-identity

$$D\Omega = 0 \tag{2.1.49}$$

and the relation

$$D^2 = \frac{i}{\hbar} \operatorname{ad} \left(\Omega\right). \tag{2.1.50}$$

Proof. The Bianchi-identity follows from

$$D\Omega = \nabla\Omega + \frac{i}{\hbar} [\gamma_D, \Omega] =$$

$$= \nabla \tilde{R} + \nabla^2 \gamma_D + \frac{i}{\hbar} [\nabla \gamma_D, \gamma_D] + \frac{i}{\hbar} [\gamma_D, \tilde{R}] + \frac{i}{\hbar} [\gamma_D, \nabla \gamma_D] + \frac{i}{\hbar} [\gamma_D, \gamma_D^2].$$
(2.1.51)

By the Bianchi identity for ∇ the first term vanishes, the last one as γ_D commutes with γ_D^2 . By Eq. (2.1.39) the second and the fourth term cancel each other, hence $D\Omega = 0$. Using Proposition 2.1.10 the second equation follows immediately:

$$D^{2} = \nabla^{2} + \frac{i}{\hbar} \operatorname{ad} (\nabla \gamma_{D}) + \frac{1}{2} \left(\frac{i}{\hbar}\right)^{2} \operatorname{ad} ([\gamma_{D}, \gamma_{D}])$$

$$= \frac{i}{\hbar} \operatorname{ad} \left(\tilde{R} + \nabla \gamma_{D} + \frac{i}{\hbar} \gamma_{D} * \gamma_{D}\right).$$

(2.1.52)

We now will look for Abelian D or in other words for conditions on γ_D which guarantee D to be Abelian. To achieve this let us write γ_D in the form

$$\gamma_D = \omega + r, \tag{2.1.53}$$

where $r \in \Omega^1 AM$. Then we have

$$\Omega = \tilde{R} + \nabla r + \frac{i}{\hbar} r * r + \frac{i}{\hbar} \operatorname{ad} (\omega)(r) - 1 \otimes \omega, \qquad (2.1.54)$$

as $\omega * \omega = i\hbar 1 \otimes \omega$. If now r fulfills

$$\delta(r) = \tilde{R} + \nabla r + \frac{i}{\hbar}r * r, \qquad (2.1.55)$$

then $\Omega = -1 \otimes \omega$, hence D will be Abelian.

2.1.12 Lemma An element $r \in \Omega^1 AM$ with $\deg_F r \ge 2$ fulfills $\delta^- r = 0$ and Eq. (2.1.55) if and only if

$$r = \delta^{-}\tilde{R} + \delta^{-} \left(\nabla r + \frac{i}{\hbar}r * r\right).$$
(2.1.56)

Proof. If the first condition is satisfied, (2.1.56) follows easily from $(\delta^-\delta + \delta\delta^-)r = r$. Let us show the converse and suppose (2.1.56) to be true. Then obviously $\delta^-r = 0$ by $(\delta^-)^2 = 0$. Let D be the Weyl connection on ΩAM with $\gamma_D = \omega + r$. To prove (2.1.55) it then suffices to show $\Omega = -1 \otimes \omega$. We have

$$\delta^{-}(\Omega+1\otimes\omega) = \delta^{-}\left(\tilde{R}+\nabla r+\frac{i}{\hbar}r*r\right) - \delta^{-}\delta r = r - \delta^{-}\delta r = \delta\delta^{-}r = 0, \qquad (2.1.57)$$

hence by the Bianchi identity $D\Omega = 0$ and $D(1 \otimes \omega) = 1 \otimes d\omega = 0$ the relation

$$\delta(\Omega + 1 \otimes \omega) = (D + \delta)(\Omega + 1 \otimes \omega) \tag{2.1.58}$$

is true. Using the Hodge-deRham decomposition in $\Omega {\rm A}M$ this entails

$$\Omega + 1 \otimes \omega = \delta^{-}(D + \delta)(\Omega + 1 \otimes \omega) = \delta^{-}\left(\nabla + \frac{i}{\hbar} \operatorname{ad}(r)\right)(\Omega + 1 \otimes \omega).$$
(2.1.59)

As the operator $\delta^{-}\left(\nabla + \frac{i}{\hbar} \operatorname{ad}(r)\right)$ raises the F-degree by 1, we must have $\Omega + 1 \otimes \omega = 0$. But this gives the claim.

II.3. Quantum spin systems

3.1. The quasi-local algebra of a spin lattice model

3.1.1 By a *Bravais lattice* or briefly just a *lattice* one understands a subgroup Λ of the additive group \mathbb{R}^d of the form

$$\Lambda = \left\{ \sum_{i=1}^{d} \lambda_i a_i \mid \lambda_i \in \mathbb{Z} \text{ for } i = 1, \dots, d \right\} ,$$

where (a_1, \ldots, a_d) is a basis of \mathbb{R}^d . We then say that Λ is the lattice *induced* by the basis (a_1, \ldots, a_d) . The length d of an inducing basis will be called the *dimension* of the lattice. Note that the dimension is uniquely determined by a given lattice but that there might be several bases by which the lattice is induced. The lattice \mathbb{Z}^d will be called the *standard* or *cubic lattice* in dimension d. It is induced by the standard basis (e_1, \ldots, e_d) of \mathbb{R}^d .

3.1.2 Let Λ be a lattice of dimension d, and denote by $\mathcal{P}_{fin}(\Lambda)$ the set of all finite subsets of Λ . Fix a natural number $N \ge 1$ and call $\frac{N}{2}$ the *spin degree* of the spin lattice model we are going to define. For each $x \in \Lambda$ let \mathcal{H}_x be the N + 1-dimensional complex Hilbert space \mathbb{C}^{N+1} . Now put for $\mathcal{O} \in \mathcal{P}_{fin}(\Lambda)$

$$\mathcal{H}_{\mathcal{O}} = \bigotimes_{x \in \mathcal{O}} \mathcal{H}_{x}$$

and define the local algebra over O as the C^* -algebra

$$\mathfrak{A}_{\mathbb{O}} = \mathfrak{B}(\mathcal{H}_{\mathbb{O}})$$
 .

Note that due to their finite dimensionality the tensor product of finitely many Hilbert spaces \mathcal{H}_x coincides here with their Hilbert tensor product. If \mathcal{O}_1 and \mathcal{O}_2 are two finite subsets of λ such that \mathcal{O}_1 is a subset of \mathcal{O}_2 , then one has the natural embedding

$$\alpha_{\mathcal{O}_1,\mathcal{O}_2}:\mathfrak{A}_{\mathcal{O}_1}\hookrightarrow\mathfrak{A}_{\mathcal{O}_2}$$

which, under the natural identification $\mathfrak{A}_{\mathbb{O}} \cong \bigotimes_{x \in \mathbb{O}} \mathfrak{B}(\mathcal{H}_x)$, maps a tensor of the form $\bigotimes_{x \in \mathbb{O}_1} A_x$ with $A_x \in \mathfrak{B}(\mathcal{H}_x)$ for all $x \in \mathbb{O}_1$ to the simple tensor $\bigotimes_{x \in \mathbb{O}_2} A_x$, where A_x is defined to be $\mathbb{1}_{\mathcal{H}_x}$ whenever $x \in \mathbb{O}_2 \setminus \mathbb{O}_1$. In more abstract terms, $\alpha_{\mathbb{O}_1,\mathbb{O}_2}$ is the unique linear map making the diagram

commute where $\pi_0 : \prod_{x \in \mathcal{O}} \mathfrak{A}_x \to \bigotimes_{x \in \mathcal{O}} \mathfrak{A}_x$ is the canonical projection mapping the family $(A_x)_{x \in \mathcal{O}}$ to $\bigotimes_{x \in \mathcal{O}} A_x$ and $\overline{\alpha}_{0_1, 0_2}$ is the map

$$\overline{\alpha}_{\mathcal{O}_1,\mathcal{O}_2}:\prod_{x\in\mathcal{O}_1}\mathfrak{A}_x\to\bigotimes_{x\in\mathcal{O}_2}\mathfrak{A}_x,\ (A_x)_{x\in\mathcal{O}_1}\mapsto(\otimes_{x\in\mathcal{O}_1}A_x)\otimes(\otimes_{x\in\mathcal{O}_2\setminus\mathcal{O}_1}\mathbb{1}_{\mathcal{H}_x})$$

II.4. Molecular quantum mechanics

4.1. The von Neumann–Wigner no-crossing rule

4.1.1 Theorem (von Neumann & Wigner (1929)) For any positive integer n let

 $\mathfrak{Herm}(n) = \left\{ A \in \mathfrak{gl}(n, \mathbb{C}) \mid A^* = A \right\}$

be the space of all (complex) hermitian $n \times n$ matrices and

$$\mathfrak{Sym}(n) = \{ A \in \mathfrak{gl}(n, \mathbb{R}) \mid A^{\mathrm{t}} = A \}$$

the space of all (real) symmetric $n \times n$ matrices. Then $\mathfrak{Herm}(n)$ and $\mathfrak{Sym}(n)$ are real vector space of dimension n^2 and $\frac{n(n+1)}{2}$, respectively. The subspaces $\mathfrak{Herm}_{dgt}(n) \subset \mathfrak{Herm}(n)$ and $\mathfrak{Sym}_{dgt}(n) \subset \mathfrak{Sym}(n)$ of hermitian respectively symmetric $n \times n$ matrices having at least one degenerate eigenvalue are (real) algebraic varieties of codimension 3 and 2, respectively.

4.1.2 Remark Recall that an eigenvalue of a real or complex $n \times n$ matrix is called *degenerate* if its algebraic multiplicity is at least 2. For hermitian or symmetric matrices this is equivalent to the geometric multiplicity of the eigenvalue being ≥ 2 .

Proof. Since the diagonal elements of a hermitian matrix $A = (a_{ij})_{1 \le i,j \le n}$ are all real and $a_{ij} = \overline{a_{ji}}$ for $i \ne j$, the (real) dimension of $\mathfrak{Herm}(n)$ is given as the sum of the number of diagonal elements of A and twice the number of its upper diagonal elements. So one obtains

dim
$$\mathfrak{Herm}(n) = n + 2 \sum_{k=1}^{n-1} k = n + (n-1)n = n^2$$

In the real symmetric case, one needs to count the number of diagonal or upper diagonal elements, hence

$$\dim \mathfrak{Sym}(n) = \sum_{k=1}^{n} k = \frac{n(n+1)}{2} .$$

The eigenvalues of a complex hermitian or real symmetric matrix A coincide with the zeros of its characteristic polynomial $\chi_A = \det(A - \lambda I_n) \in \mathbb{C}[\lambda]$. Let $\mathbb{D}(\chi_A)$ be the discriminant of the characteristic polynomial; see (Cohen, 1993, Sec. 3.3.2) for the definition and properties of the discriminant. Then $\mathbb{D}(\chi_A)$ is a polynomial in the coefficients of χ_A and vanishes if and only if χ_A has a multiple root. Since the coefficients of χ_A are polynomials in the entries of A, the set of hermitian (respectively symmetric) $n \times n$ matrices with a degenerate eigenvalue is a real algebraic variety in $\mathfrak{Herm}(n)$ (respectively $\mathfrak{Spm}(n)$). Next let us determine the codimension of the variety $\mathfrak{Herm}_{dgt}(n)$. To this end recall that a hermitian matrix A can be written in the form $A = UDU^{-1}$, where D is a diagonal matrix having the eigenvalues of A as its entries and where U is a complex unitary $n \times n$ matrix. The diagonal matrix $D = (d_{ij})_{1 \leq i,j \leq n}$ is uniquely determined when one requires that its diagonal entries are linearly ordered so that $d_{11} \leq \ldots \leq d_{nn}$. The matrix U is uniquely up to a unitary matrix V commuting with D. In case A has n different eigenvalues, the only unitary matrices commuting with D are diagonal matrices with entries from U(1). Since dim U(n) = n^2 Hence the codimension of $\mathfrak{Herm}_{dgt}(n)$ in $\mathfrak{Herm}(n)$ is

Part III. Quantum Field Theory

III.1. Representations of the Lorentz and Poincaré groups

1.1. The Lorentz invariant measure on a mass hyperboloid

1.1.1 Consider Minkowski space of space dimension d that is \mathbb{R}^{1+d} endowed with the Minkowski inner product

$$\langle \cdot, \cdot \rangle_{\mathsf{M}} : \mathbb{R}^{1+d} \times \mathbb{R}^{1+d} \to \mathbb{R}, \ (p,q) \mapsto -p^0 q^0 + \langle \vec{p}, \vec{q} \rangle = p^0 q^0 - \sum_{i=1}^d p^i q^i$$

Note that $\langle \cdot, \cdot \rangle$ stands here for the euclidean inner product, and \vec{p} is the *spacial vector* (p^1, \ldots, p^d) associated to the *space-time vector* $p \in \mathbb{R}^{1+d}$. We sometimes will denote space-time dimension 1 + d by D. For m > 0 let

$$H_m^+ = \{ p \in \mathbb{R}^D \mid \langle p, p \rangle_{\mathsf{M}} = m^2 \& p^0 > 0 \}$$

be the positive mass hyperboloid of mass m. Observe that

$$\chi^+: \mathbb{R}^d \to H_m^+, \ \mathsf{p} \mapsto \big(E(\mathsf{p}),\mathsf{p}\big) \quad \text{with} \ E(\mathsf{p}) = \sqrt{m^2 + \langle \mathsf{p},\mathsf{p}\,\rangle}$$

is a global chart of the mass hyperboloid. Its inverse is given by

$$\vec{}: H_m^+ \to \mathbb{R}^d, \ p = (p^0, p^1, \dots, p^d) \mapsto \vec{p} = (p^1, \dots, p^d) \ .$$

Note that $E(\vec{p}) = p^0$ for all $p \in H_m^+$.

Now let λ denote Lebesgue measure on \mathbb{R}^d . We will show that the pushforward measure $\Omega_m = \chi^+_* \left(\frac{1}{E}\lambda\right)$ is a Lorentz invariant measure on H^+_m that is $\Lambda_*\Omega_m = \Omega_m$ for all $\Lambda \in \mathsf{SO}^\uparrow(1,d)$. Note that we have used here that Λ leaves H^+_m invariant.

1.1.2 Lemma For $\Lambda \in SO^{\uparrow}(1, d)$ let Ψ_{Λ} denote the map

$$\Psi_{\Lambda}: \mathbb{R}^d \to \mathbb{R}^d, \ \mathsf{p} \mapsto \Psi_{\Lambda}(\mathsf{p}) = \overline{\Lambda \chi^+(\mathsf{p})} \ .$$

Then Ψ_{Λ} is a diffeomorphism and the following holds true:

(i) The map $SO^{\uparrow}(1,d) \rightarrow Diff(\mathbb{R}^d)$, $\Psi : \Lambda \mapsto \Psi_{\Lambda}$ is a homomorphism that is

$$\Psi_{\Lambda_1\Lambda_2} = \Psi_{\Lambda_1}\Psi_{\Lambda_2}$$
 for all $\Lambda_1, \Lambda_2 \in \mathsf{SO}^{\uparrow}(1, d)$.

(ii) The jacobian of Ψ_{Λ} is given by

$$J_{\Psi_{\Lambda}} = |D\Psi_{\Lambda}| = \det \circ D\Psi_{\Lambda} = \frac{E \circ \Psi_{\Lambda}}{E}$$
.

Proof. ad (i). Let $\Lambda_1, \Lambda_2 \in \mathsf{SO}^{\uparrow}(1, d)$, $\mathsf{p} \in \mathbb{R}^d$ and compute

$$\Psi_{\Lambda_1}\Psi_{\Lambda_2}(\mathsf{p}) = \Psi_{\Lambda_1}\left(\overrightarrow{\Lambda_2\chi^+(\mathsf{p})}\right) = \overrightarrow{\Lambda_1\Lambda_2\chi^+(\mathsf{p})} = \Psi_{\Lambda_1\Lambda_2}(\mathsf{p}) \ .$$

This implies in particular that Ψ_{Λ} is a diffeomorphism with inverse $\Psi_{\Lambda^{-1}}$.

ad (ii). Assume first that $\Lambda \in \mathsf{SO}^{\uparrow}(1,d)$ is a rotation that is $\Lambda = \begin{pmatrix} 1 & 0 \\ 0 & R \end{pmatrix}$ for some $R \in \mathsf{SO}(d)$. Then observe that $\Psi_{\Lambda} = R$ and compute for $\mathsf{p} \in \mathbb{R}^d$

$$E(\Psi_{\Lambda}\mathsf{p}) = E(R\mathsf{p}) = \sqrt{m^2 + \langle R\mathsf{p}, R\mathsf{p} \rangle} = \sqrt{m^2 + \langle \mathsf{p}, \mathsf{p} \rangle} = E(\mathsf{p}) \;.$$

Hence

$$\det \left(D\Psi_{\Lambda}(\mathbf{p}) \right) = 1 = \frac{E(\Psi_{\Lambda}\mathbf{p})}{E(\mathbf{p})}$$

Next let Λ be a Lorentz boost in the direction of p^1 that is let $\Lambda = \begin{pmatrix} \cosh \tau & \sinh \tau & 0 \\ \sinh \tau & \cosh \tau & 0 \\ 0 & 0 & 1 \end{pmatrix}$ where $\tau \in \mathbb{R}$ and 1 denotes the identity matrix over \mathbb{R}^{d-1} .

Then compute with $\Psi_{\Lambda,i}$ for $i = 1, \ldots, d$ denoting the *i*-th component of $\Psi_{\Lambda} : \mathbb{R}^d \to \mathbb{R}^d$:

$$\Psi_{\Lambda,i}(\mathbf{p}) = \Lambda_{i0}E(\mathbf{p}) + \sum_{i=1}^{d} \Lambda_{ij}\mathbf{p}^{j} = \begin{cases} \sinh \tau \cdot E(\mathbf{p}) + \cosh \tau \cdot \mathbf{p}^{1} & \text{for } i = 1, \\ \mathbf{p}^{i} & \text{for } i = 2, \dots, d, \end{cases}$$
$$\frac{\partial \Psi_{\Lambda,i}}{\partial \mathbf{p}^{j}}(\mathbf{p}) = \begin{cases} \sinh \tau \cdot \frac{\mathbf{p}^{1}}{E(\mathbf{p})} + \cosh \tau & \text{for } i = j = 1, \\ \sinh \tau \cdot \frac{\mathbf{p}^{j}}{E(\mathbf{p})} & \text{for } i = 1 \text{ and } j = 2, \dots, d, \\ 0 & \text{for } i = 2, \dots, d, \\ \delta_{ij} & \text{for } i = 2, \dots, d, \end{cases}$$

and

$$E^{2}(\Psi_{\Lambda}\mathsf{p}) = \sinh^{2}\tau \cdot E^{2}(\mathsf{p}) + 2\sinh\tau\cosh\tau \cdot E(\mathsf{p})\cdot\mathsf{p}^{1} + \cosh^{2}\tau \cdot (\mathsf{p}^{1})^{2} + \sum_{i=2}^{a} (\mathsf{p}^{i})^{2} + m^{2} =$$

= $(\sinh^{2}\tau + 1)\cdot E^{2}(\mathsf{p}) + 2\sinh\tau\cosh\tau \cdot E(\mathsf{p})\cdot\mathsf{p}^{1} + (\cosh^{2}\tau - 1)\cdot(\mathsf{p}^{1})^{2} =$
= $(\cosh\tau\cdot E(\mathsf{p}) + \sinh\tau\cdot\mathsf{p}^{1})^{2}$.

This entails the equality

$$\det \left(D\Psi_{\Lambda}(\mathbf{p}) \right) = \sinh \tau \cdot \frac{\mathbf{p}^{1}}{E(\mathbf{p})} + \cosh \tau = \frac{E(\Psi_{\Lambda}\mathbf{p})}{E(\mathbf{p})}$$

Since $SO^{\uparrow}(1,d)$ is generated by the rotations and Lorentz boosts in direction p^1 and since by (i)

$$\det \left(D\Psi_{\Lambda_1\Lambda_2}(\mathsf{p}) \right) = \det \left(D\Psi_{\Lambda_1}(\Psi_{\Lambda_2}\mathsf{p}) \right) \cdot \det \left(D\Psi_{\Lambda_2}(\mathsf{p}) \right)$$

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the claim follows.

1.1.3 Proposition With notation as above the pushforward measure $\Omega_m = \chi^+_* \left(\frac{1}{\omega}\lambda\right)$ is a Lorentz invariant measure on the positive mass hyperboloid H_m^+ that is

$$\int_{H_m^+} f(\Lambda p) \, d\Omega_m(p) = \int_{H_m^+} f(p) \, d\Omega_m(p) \tag{1.1.1}$$

for all $f \in L^1(H_m^+, \Omega_m)$ and $\Lambda \in \mathrm{SO}^{\uparrow}(1, d)$.

Proof. By definition of the pushforward measure Ω_m is the unique Borel measure on H_m^+ such that for all $f \in \mathcal{C}_{cpt}(H_m^+)$

$$\int_{H_m^+} f(p) \, d\Omega_m(p) = \int_{\mathbb{R}^d} f(\chi^+ \mathsf{p}) \, \frac{1}{E(\mathsf{p})} d\lambda(\mathsf{p}) \, .$$

The claim follows from this observation since for all $\Lambda \in \mathsf{SO}^{\uparrow}(1,d)$ the equality

$$\begin{split} \int_{\mathbb{R}^d} f(\Lambda \chi^+ \mathsf{p}) \frac{1}{E(\mathsf{p})} d\lambda(\mathsf{p}) &= \int_{\mathbb{R}^d} f(\chi^+ \Psi_\Lambda \mathsf{p}) \frac{1}{E(\mathsf{p})} d\lambda(\mathsf{p}) = \\ &= \int_{\mathbb{R}^d} f(\chi^+ \Psi_\Lambda \mathsf{p}) \frac{1}{E(\Psi_\Lambda \mathsf{p})} \det \left(D\Psi_\Lambda(\mathsf{p}) \right) d\lambda(\mathsf{p}) = \int_{\mathbb{R}^d} f(\chi^+ \mathsf{p}) \frac{1}{E(\mathsf{p})} d\lambda(\mathsf{p}) \ . \end{split}$$

holds true by Lemma 1.1.2 (ii) .

III.2. Axiomatic quantum field theory à la Wightman and Gårding

2.1. Wightman axioms

2.1.1 The Wightman axioms were first introduced in the paper Wightman & Gårding (1964), and then explained in more detail in the textbooks Jost (1965) and (Streater & Wightman, 2000, Sec. 3.1). The latter is still the main reference for the axiomatic treatment of quantum field theory in the spirit of Wightman and Gårding. See also (Schottenloher, 2008, Sec. 8.3) for a more modern formulation which we follow here.

2.1.2 Definition A Wightman quantum field theory of space-time dimension D = d + 1, $d \in \mathbb{N}_{>0}$, consists of the following data:

- the state space of the theory given by the projective space P(H) associated to a separable complex Hilbert space H,
- a distinguished state $\omega_{\circ} = \mathbb{C}v_{\circ} \in \mathbb{P}(\mathcal{H})$ called the *vacuum state* together with the choice of a normalized representing vector $v_{\circ} \in \mathcal{H}$ called *vacuum vector*,
- a unitary representation $U: \widetilde{\mathsf{P}^{\uparrow}_+}(d+1) \to \mathsf{U}(\mathcal{H})$ of the universal cover

$$\widetilde{\mathsf{P}_+^{\uparrow}}(d+1) \cong \mathbb{R}^{d+1} \rtimes \widetilde{\mathsf{SO}^{\uparrow}}(1,d)$$

of the proper orthochronous Poincaré group $\mathsf{P}^{\uparrow}_+(d+1) = \mathbb{R}^{1+d} \rtimes \mathsf{SO}^{\uparrow}(1,d)$,

• and finally a family $(\Phi^j)_{1 \leq j \leq n}$, $n \in \mathbb{N}_{>0}$, of so-called *field operators*

$$\Phi^j: \mathcal{S}(\mathbb{R}^{d+1}) \to \mathfrak{L}_{\mathsf{u}}(\mathcal{H})$$

which are defined on the Schwartz space of rapidly decreasing functions on \mathbb{R}^n and map to the space of unbounded linear operators on the Hilbert space \mathcal{H} .

These data are assumed to fulfill the following axioms, the so-called Wightman axioms:

(W1) (Assumptions about the domain and the continuity of the field)

There exists a dense linear subspace $\mathcal{D} \subset \mathcal{H}$ containing v_{\circ} such that \mathcal{D} is contained in the domain of all the operators $\Phi^{j}(f)$ and their adjoints $\Phi^{j}(f)^{*}$, where $f \in \mathcal{S}(\mathbb{R}^{d+1})$ and $j = 1, \ldots, k$. Moreover, the unitary representation U and the operators $\Phi^{j}(f)$ and $\Phi^{j}(f)^{*}$ leave \mathcal{D} invariant that is

$$U(a, A)\mathcal{D} \subset \mathcal{D}, \quad \Phi^{j}(f)\mathcal{D} \subset \mathcal{D}, \ \Phi^{j}(f)^{*}\mathcal{D} \subset \mathcal{D}$$

for all $(a, A) \in \widetilde{\mathsf{P}_+^{\uparrow}}(1, d)$, $f \in S(\mathbb{R}^{d+1})$ and $j = 1, \ldots, k$. Finally, for every $v \in \mathcal{D}$, $w \in \mathcal{H}$ and $j = 1, \ldots, n$ the maps

$$\mathfrak{S}(\mathbb{R}^{d+1}) \to \mathbb{C}, \ f \mapsto \langle w, \Phi^j(f)v \rangle$$

are tempered distributions.

(W2) (*Transformation law of the field*) For all $(a, A) \in \widetilde{\mathsf{P}_+^{\uparrow}}(d+1)$ and all $f \in S(\mathbb{R}^{d+1})$ the equation

$$U(a,A)\Phi^{j}(f)U(a,A)^{-1} = \sum_{k=1}^{n} \varrho^{jk} (A^{-1})\Phi^{k}((a,A)f)$$

is valid over the domain D, where $\varrho : \widetilde{SO}^{\uparrow}(1, d) \to GL(n, \mathbb{C})$ is a finite dimensional representation of the universal cover of the proper orthochronous Lorentz group $SO^{\uparrow}(1, d)$ and the action of $\widetilde{P}(d+1)$ on $S(\mathbb{R}^{d+1})$ is given by

$$\widetilde{\mathsf{P}}(d+1) \times \mathscr{S}(\mathbb{R}^{d+1}) \to \mathscr{S}(\mathbb{R}^{d+1}),$$
$$((a,A), f) \mapsto (a,A)f = \left(\mathbb{R}^{d+1} \ni x \mapsto f(A^{-1}(x-a)) \in \mathbb{C}\right).$$

(W3) (Local commutativity or microscopic causality)

If the support of test functions $f, g \in S(\mathbb{R}^{d+1})$ is space-like separated that is if f(x) g(y) = 0 for all $x, y \in \mathbb{R}^{d+1}$ with $\langle x - y, x - y \rangle_{\mathsf{M}} \ge 0$, then for all $j, k = 1, \dots, n$ the relation

$$[\Phi^{j}(f), \Phi^{k}(g)]_{-} = [\Phi^{j}(f), \Phi^{j}(g)^{*}]_{-} = 0$$

or the relation

$$[\Phi^{j}(f), \Phi^{k}(g)]_{+} = [\Phi^{j}(f), \Phi^{j}(g)^{*}]_{+} = 0$$

holds true over the domain \mathcal{D} . Hereby, $[S,T]_{-}$ denotes the *commutator*

 $[S,T]_{-}: \mathcal{D} \to \mathcal{H}, \quad v \mapsto STv - TSv$

and $[S,T]_+$ the anti-commutator

$$[S,T]_+: \mathcal{D} \to \mathcal{H}, \quad v \mapsto STv + TSv$$

of two operators $S, T \in \mathfrak{L}_{u}(\mathcal{H})$ which are both assumed to be defined on the domain \mathcal{D} and to leave it invariant.

(W4) (Cyclicity of the vacuum vector)

The linear span of the set of all elements $v \in \mathcal{H}$ of the form

$$v = \Phi^{j_1}(f_1) \dots \Phi^{j_m}(f_m) v_\circ$$

where $m \in \mathbb{N}$, $1 \leq j_1, \ldots, j_m \leq n$, and $f_1, \ldots, f_m \in \mathcal{S}(\mathbb{R}^{d+1})$, is dense in \mathcal{H} .

2.1.3 Remarks (a) The vacuum vector v_{\circ} being normalized just means that $||v_{\circ}|| = 1$. This implies that the vacuum state ω_{\circ} determines v_{\circ} only up to a factor $z \in S^1 \subset \mathbb{C}$. The physically measurable quantities of the quantum field theory such as expectation values or transition amplitudes do not depend on that choice.

(b) The field operators Φ^j are operator valued distributions. This reflects the fact that only the "smeared" fields $\Phi^j(f)$ can be interpreted physically as observable. The notation $\Phi^j(x)$ for a field evaluated at a space-time point $x \in \mathbb{R}^{1,3}$ therefore does not make sense, neither mathematically nor physically. Nevertheless it is often used for reasons of convenience, in particular in the physics literature. The smeared field $\Phi^j(f)$ then is interpreted, again imprecisely, as the integral

$$\Phi^j(f) = \int_{\mathbb{R}^{d+1}} f(x) \, \Phi^j(x) \, dx \; .$$

We will avoid the notation of pointwise evaluated fields in the formulation of definitions and theorems, but occasionally use it as a heuristic.

For example, Axiom (W3) can heuristically be interpreted as saying that the (anti-) commutation relations

$$[\Phi^{j}(x), \Phi^{k}(y)]_{\mp} = [\Phi^{j}(x), \Phi^{j}(y)^{*}]_{\mp} = 0$$

hold true for $x, y \in \mathbb{R}^{1,d}$ space-like separated which means for the situation when

$$\langle x-y, x-y \rangle_{\mathsf{M}} < 0$$
.

2.2. Fock space

2.2.1 Recall from Section A.3.4 that the Hilbert tensor product $\mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$ of two Hilbert spaces \mathcal{H}_1 and \mathcal{H}_2 is defined as the completion of the the algebraic tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$ endowed with the inner product

$$\langle \cdot, \cdot \rangle : (\mathcal{H}_1 \otimes \mathcal{H}_2) \times (\mathcal{H}_1 \otimes \mathcal{H}_2) \to \mathbb{K}, (v_1 \otimes v_2, w_1 \otimes w_2) \mapsto \langle v_1, w_1 \rangle \cdot \langle v_2, w_2 \rangle.$$

The norm of an element $v_1 \otimes v_2 \in \mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$ is then given by $||v_1 \otimes v_2|| = ||v_1|| \cdot ||v_2||$, and every element $v \in \mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$ can be written as the sum of a square summable family $(v_{i1} \otimes v_{i2})_{i \in I}$ that is as

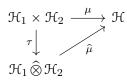
$$v = \sum_{i \in I} v_{i1} \otimes v_{i2}$$
 where $||v||^2 = \sum_{i \in I} ||v_{i1}||^2 \cdot ||v_{i2}||^2 < \infty$.

If $(e_i)_{i\in I}$ is Hilbert basis for \mathcal{H}_1 and $(f_j)_{j\in J}$ one of \mathcal{H}_2 , the family $(e_i \otimes f_j)_{(i,j)\in I\times J}$ is a Hilbert basis of $\mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$. Moreover, the canonical map $\tau : \mathcal{H}_1 \times \mathcal{H}_2 \to \mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$, $(v_1, v_2) \mapsto v_1 \otimes v_2$ is bilinear and weakly Hilbert–Schmidt that means that there exists a $C \ge 0$ such that for all Hilbert bases $(e_i)_{i\in I}$ of \mathcal{H}_1 , all Hilbert bases $(f_j)_{j\in J}$ of \mathcal{H}_2 , and all $w \in \mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$

$$\sum_{(i,j)\in I\times J} |\langle \tau(e_i, f_j), w \rangle|^2 \leq C ||w||^2$$

Note that if this condition holds for one Hilbert basis of \mathcal{H}_1 and one of \mathcal{H}_2 , it holds for all. The Hilbert tensor product, which in the following we will only call tensor product, satisfies the following universal property.

(HTensor) For every Hilbert space \mathcal{H} and every weakly Hilbert–Schmidt bilinear map $\mu : \mathcal{H}_1 \times \mathcal{H}_2 \rightarrow \mathcal{H}$ there exists a unique bounded linear map $\hat{\mu} : \mathcal{H}_1 \otimes \mathcal{H}_2 \rightarrow \mathcal{H}$ such that the diagram



commutes.

For a proof of the universal property see Section A.3.4 or (Kadison & Ringrose, 1997, Sec. 2.6.). Note that by its universal property the Hilbert tensor product $\hat{\otimes}$ is a bifunctor on the category Hilb of Hilbert spaces and bounded maps. Moreover, Hilb equipped with the bifunctor $\hat{\otimes}$ becomes a monoidal category. See Section A.3.4 for details and proofs.

2.2.2 Now let us fix a Hilbert space \mathcal{H} and consider the higher Hilbert tensor product powers $\mathfrak{F}^n(\mathcal{H}) = \mathcal{H}^{\hat{\otimes} n}$ for natural n. These are recursively defined by

$$\mathfrak{H}^{\widehat{\otimes} 0} = \mathbb{K}, \quad \mathfrak{H}^{\widehat{\otimes} n+1} = \mathfrak{H}\widehat{\otimes} (\mathfrak{H}^{\widehat{\otimes} n})$$

The Fock space of $\mathcal H$ now is defined as the Hilbert space direct sum

$$\mathfrak{F}(\mathcal{H}) = \widehat{\bigoplus}_{n \in \mathbb{N}} \mathfrak{F}^n(\mathcal{H}) = \widehat{\bigoplus}_{n \in \mathbb{N}} \mathcal{H}^{\widehat{\otimes} n}$$

Its elements are families $(v_n)_{n\in\mathbb{N}}$ of vectors $v_n \in \mathcal{H}^{\widehat{\otimes} n}$ such that $\sum_{n\in\mathbb{N}} \|v_n\|^2 < \infty$. The inner product of two such families $v = (v_n)_{n\in\mathbb{N}}, w = (w_n)_{n\in\mathbb{N}} \in \mathfrak{F}(\mathcal{H})$ is given, according to ??, by

$$\langle v, w \rangle = \sum_{n \in \mathbb{N}} \langle v_n, w_n \rangle .$$

2.2.3 Remark The construction of the Fock space resembles the one of the tensor algebra. Recall that the tensor algebra of \mathcal{H} is the vector space $T(\mathcal{H}) = \bigoplus_{n \in \mathbb{N}} T^n(\mathcal{H})$ where $T^n(\mathcal{H})$ is defined as the algebraic tensor product power $\mathcal{H}^{\otimes n}$. The *completed tensor algebra* of \mathcal{H} now is the ℓ^1 -completion

$$\widehat{\mathsf{T}}(\mathcal{H}) = \ell_1 - \bigoplus_{n \in \mathbb{N}} \widehat{\mathsf{T}}^n(\mathcal{H}) ,$$

where $\widehat{\mathsf{T}}^n(\mathcal{H}) = \mathfrak{F}^n(\mathcal{H}) = \mathcal{H}^{\hat{\otimes} n}$. The completed tensor algebra lies densely in Fock space. To verify this observe that, regarded in the category of Banach spaces, Fock space (including its norm) coincides with the ℓ_2 -direct sum of the spaces Banach spaces $\widehat{\mathsf{T}}^n(\mathcal{H})$ and $\widehat{\mathsf{T}}(\mathcal{H})$ with their ℓ_1 -direct sum. Since for every summable family $v = (v_n)_{\in \mathbb{N}}$ with $v_n \in \widehat{\mathsf{T}}^n(\mathcal{H})$ the relation

$$\|v\| = \|v\|_2 = \sqrt{\sum_{n \in \mathbb{N}} \|v_n\|^2} \le \sqrt{\sum_{n \in \mathbb{N}} \|v_n\|} \cdot \sqrt{\sup_{n \in \mathbb{N}} \|v_n\|} \le \sum_{n \in \mathbb{N}} \|v_n\| = \|v\|_1$$

holds true by Hölders inequality for series, $\widehat{\mathsf{T}}(\mathcal{H})$ is contained in $\mathfrak{F}(\mathcal{H})$. It is also dense in Fock space because the (algebraic) direct sum $\bigoplus_{n\in\mathbb{N}}\widehat{\mathsf{T}}^n(\mathcal{H})$ is already so.

Unlike Fock space in the case $\dim \mathcal{H} = \infty$, the completed tensor algebra $\widehat{T}(\mathcal{H})$ always carries a canonical algebra structure. To define the product of two summable families $v = (v_n)_{n \in \mathbb{N}}$ and $w = (w_n)_{n \in \mathbb{N}}$ one puts for all natural n

$$z_n = \sum_{k=0}^n v_k \otimes w_{n-k} \; .$$

Then $z_n \in \mathsf{T}^n(\mathcal{H})$ for all $n \in \mathbb{N}$, and the family $z = (z_n)_{n \in \mathbb{N}}$ is absolutely summable again since

$$\sum_{n \in \mathbb{N}} \|z_n\| = \lim_{N \to \infty} \sum_{n=0}^N \|z_n\| \le \lim_{N \to \infty} \sum_{n=0}^N \sum_{k=0}^n \|v_k\| \|w_{n-k}\| \le \lim_{N \to \infty} \sum_{k=0}^N \sum_{l=0}^N \|v_k\| \|w_l\| \le \|v\|_1 \|w\|_1 \ .$$

Hence $z = (z_n)_{n \in \mathbb{N}}$ is an element $\widehat{\mathsf{T}}(\mathcal{H})$ which we call the *product* of v and w. It will be denoted by $v \otimes w$. By the preceding estimate we thus obtain a continuous map

$$\otimes: \widehat{\mathsf{T}}(\mathcal{H}) \times \widehat{\mathsf{T}}(\mathcal{H}) \to \widehat{\mathsf{T}}(\mathcal{H}), \ (v,w) \mapsto v \otimes w$$

such that

$$\|v \otimes w\|_1 \leqslant \|v\|_1 \|w\|_1$$
 for all $v, w \in \widehat{\mathsf{T}}(\mathcal{H})$.

The restriction of \otimes to the (uncompleted) tensor algebra $T(\mathcal{H}) = \bigoplus_{n \in \mathbb{N}} T^n(\mathcal{H})$ is associative, so by density one concludes that \otimes on $\widehat{T}(\mathcal{H})$ is associative as well. Hence $\widehat{T}(\mathcal{H})$ is a Banach algebra.

Even though $\mathfrak{F}(\mathcal{H})$ might not possess a compatible Banach algebra structure, it carries the structure of a $\widehat{\mathsf{T}}(\mathcal{H})$ left and right module with the left and right actions being continuous. Let us show this for the left module structure in some more detail. The right module case is analogous. So assume $v = (v_n)_{n \in \mathbb{N}} \in \widehat{\mathsf{T}}(\mathcal{H})$, $w = (w_n)_{n \in \mathbb{N}} \in \mathfrak{F}(\mathcal{H})$, and let $z = (z_n)_{n \in \mathbb{N}}$ where as before $z_n = \sum_{k=0}^n v_k \otimes w_{n-k}$. Put $w_k = 0$ for k < 0. Then compute using the triangle and Hölder's inequality

$$\begin{aligned} \|z\|_{2}^{2} &= \lim_{N \to \infty} \sum_{n=0}^{N} \|z_{n}\|^{2} = \lim_{N \to \infty} \sum_{n=0}^{N} \left\| \sum_{k=0}^{n} v_{k} \otimes w_{n-k} \right\|^{2} \leq \\ &\leq \lim_{N \to \infty} \sum_{n=0}^{N} \left(\sum_{k=0}^{N} \left(\|v_{k}\|^{1/2} \|w_{n-k}\| \right) \|v_{k}\|^{1/2} \right)^{2} \leq \\ &\leq \lim_{N \to \infty} \sum_{n=0}^{N} \left(\sum_{k=0}^{N} \|v_{k}\| \|w_{n-k}\|^{2} \right) \left(\sum_{k=0}^{N} \|v_{k}\| \right) \leq \\ &\leq \lim_{N \to \infty} \|v\|_{1} \sum_{k=0}^{N} \left(\|v_{k}\| \sum_{n=0}^{N} \|w_{n-k}\|^{2} \right) \leq \\ &\leq \lim_{N \to \infty} \|v\|_{1} \sum_{k=0}^{N} \left(\|v_{k}\| \sum_{n=0}^{N} \|w_{n}\|^{2} \right) = \|v\|_{1}^{2} \|w\|_{2}^{2}. \end{aligned}$$

Hence $z \in \mathfrak{F}(\mathfrak{H})$, and the product $\otimes : \widehat{\mathsf{T}}(\mathfrak{H}) \times \widehat{\mathsf{T}}(\mathfrak{H}) \to \widehat{\mathsf{T}}(\mathfrak{H})$ has a unique continuous extension to a left action

$$\otimes: \widehat{\mathsf{T}}(\mathcal{H}) \times \mathfrak{F}(\mathcal{H}) \to \mathfrak{F}(\mathcal{H}), \ (v, w) \mapsto v \otimes w$$

such that

$$\|v \otimes w\|_2 \leq \|v\|_1 \|w\|_2$$
 for all $v \in \widehat{\mathsf{T}}(\mathcal{H}), w \in \mathfrak{F}(\mathcal{H})$.

2.2.4 Next we will show that associating to a Hilbert space its Fock space can be extended to a functor on the category Hilb₁ of Hilbert spaces and linear contractions between them. Recall that by a linear contraction one understands a bounded linear operator with norm ≤ 1 . So assume that $A: \mathcal{H}_1 \to \mathcal{H}_2$ is a contraction between Hilbert spaces \mathcal{H}_1 and \mathcal{H}_2 . By functoriality of the algebraic tensor product one obtains for each $n \in \mathbb{N}_{>0}$ a linear map

$$A^{\otimes n}: \mathcal{H}_1^{\otimes n} \to \mathcal{H}_2^{\otimes n}, v_1 \otimes \ldots \otimes v_n \mapsto Av_1 \otimes \ldots \otimes Av_n$$

By Proposition 3.4.5 or (Kadison & Ringrose, 1997, Prop. 2.6.12 & Eq. 2.6.(16)) this operator has norm $||A||^n$ and extends uniquely to a bounded linear operator $\mathfrak{F}^n(A) : \mathfrak{F}^n(\mathcal{H}_1) \to \mathfrak{F}^n(\mathcal{H}_2)$ having the same norm. Since by assumption $||A|| \leq 1$, one concludes that $||\mathfrak{F}^n(A)|| \leq 1$ for all $n \in \mathbb{N}_{>0}$. One further puts $\mathfrak{F}^0(A) = \mathrm{id}_{\mathbb{K}}$ and observes that then $\sup_{n \in \mathbb{N}} ||\mathfrak{F}^n(A)|| = 1$. Hence, by construction of the operators $\mathfrak{F}^n(A)$ and definition of the Hilbert direct sum the map

$$\mathfrak{F}(A):\mathfrak{F}(\mathcal{H}_1)\to\mathfrak{F}(\mathcal{H}_2),\ v=(v_n)_{n\in\mathbb{N}}\mapsto(\mathfrak{F}^n(A)(v_n))_{n\in\mathbb{N}}$$

is well-defined and a bounded linear operator of norm 1. Note that hereby we have again used the (silent) agreement that $v = (v_n)_{n \in \mathbb{N}}$ denotes a square-integrable family with $v_n \in \mathfrak{F}^n(\mathcal{H}_1)$ for all $n \in \mathbb{N}$. By construction it is immediate that $\mathfrak{F}(\mathrm{id}_{\mathcal{H}}) = \mathrm{id}_{\mathfrak{F}(\mathcal{H})}$ for every Hilbert space \mathcal{H} and that for linear contractions $A : \mathcal{H}_1 \to \mathcal{H}_2$ and $B : \mathcal{H}_2 \to \mathcal{H}_3$ between Hilbert spaces the relation

$$\mathfrak{F}(BA) = \mathfrak{F}(B)\,\mathfrak{F}(A)$$

holds true. Hence we obtain as promised a (covariant) functor \mathfrak{F} from the category Hilb to itself. One sometimes calls \mathfrak{F} the *functor of second quantization*.

2.2.5 Particularly important for quantum field theory is the observation going back to Cook (1953) that every closed densely defined linear operator on a Hilbert space has an extension to Fock space which again is closed and densely defined. Let us explain this in some more detail. We essentially follow the approach by Cook (1953); see also Emch (2009).

Let $(\mathcal{H})_{i=1}^n$ be a finite family of Hilbert spaces and $(A_i)_{i=1}^n$ a family of closed densely defined unbounded linear operators $A_i : \text{Dom}(A_i) \subset \mathcal{H}_i \to \mathcal{H}_i$, $i = 1, \ldots, n$ over the same index set. Hence the adjoint A_i^* of A_i is a closed densely defined unbounded linear operator on \mathcal{H}_i for every index $i = 1, \ldots, n$. Put $\mathcal{D}_i = \text{Dom}(A_i)$ and $\mathcal{D}_i^* = \text{Dom}(A_i^*)$ and note that then \mathcal{D}_i and \mathcal{D}_i^* are dense in \mathcal{H}_i by assumption and the preceding observation.

2.3. The free scalar field

2.3.1 Here we want to show that a model of the Wightman axioms is given by the free scalar field of mass m > 0 in space-time dimension D = d + 1 for $d \in \mathbb{N}_{>0}$. The Hilbert space \mathcal{H} of the free scalar field is the symmetric Fock space $\mathfrak{F}_{s}(L^{2}(H_{m}^{+},\Omega_{m}))$ over the 1-particle Hilbert space $L^{2}(H_{m}^{+},\Omega_{m})$ of square-integrable functions on the positive mass hyperboloid $H_{m}^{+} \subset \mathbb{R}^{D}$ equipped with the lorentz-invariant measure Ω_{m} which has been defined in Section 1.1. By definition, Ω_{m} coincides with the pushforward measure $\chi_{*}^{+}(\frac{1}{E}\lambda)$, where λ denotes Lebesgue measure on \mathbb{R}^{d} , $E(\mathbf{p}) = \sqrt{m^{2} + \langle \mathbf{p}, \mathbf{p} \rangle}$ for all $\mathbf{p} \in \mathbb{R}^{d}$, and $\chi^{+} : \mathbb{R}^{d} \to H_{m}^{+}$ is the chart of the positive mass hyperpoloid which maps $\mathbf{p} \in \mathbb{R}^{d}$ to $(E(\mathbf{p}), \mathbf{p}) \in H_{m}^{+}$. In this section we will often denote the 1-particle Hilbert space of the free scalar field by $\mathcal{H}^{(1)}$.

III.3. Algebraic quantum field theory à la Haag-Kastler

3.1. The Haag-Kastler axioms

Appendix A.

Mathematical Toolbox

A.1. Topological Vector Spaces

A.1.1. Topological division rings and fields

1.1.1 Vector spaces with a compatible topology can not only defined for vector spaces over the ground fields \mathbb{R} and \mathbb{C} but also over fields \mathbb{K} carrying an absolute value $|\cdot| : \mathbb{K} \to \mathbb{R}_{\geq 0}$. This endows the ground field with a topology which will be needed in the definition of a topological vector space. We therefore give here a brief introduction to topological division rings and fields first.

1.1.2 Definition Let R be a division ring. By an *absolute value* on R one understands a map $|\cdot|: R \to \mathbb{R}_{\geq 0}$ such that the following axioms hold true.

(VDR1) The function $|\cdot|$ is multiplicative that is

$$|xy| = |x| |y|$$
 for all $x, y \in R$.

(VDR2) The triangle inequality is satisfied which means that

$$|x+y| \leq |x|+|y|$$
 for all $x, y \in R$.

(VDR3) For all $x \in R$ the relation |x| = 0 holds true if and only if x = 0.

A division ring or field endowed with an absolute value is called a *valued division ring* respectively a *valued field*. An absolute value $|\cdot|$ on a division ring R and the corresponding valued division ring $(R, |\cdot|)$ are called *non-archimedean* if the *strong triangle inequality* is satisfied that is if

(VDR4) $|x+y| \leq \max\{|x|, |y|\}$ for all $x, y \in R$.

Otherwise $|\cdot|$ and $(R, |\cdot|)$ are called *archimedean*.

1.1.3 Lemma Let $(R, |\cdot|)$ be a valued division ring. Then

(i)
$$|1| = 1$$
,

(ii) |-x| = |x| for all $x \in R$, and

(iii) $||x| - |y|| \le |x - y| \le |x| + |y|$ for all $x, y \in R$.

Proof. (i) holds true since $|1| = |1^2| = |1|^2$ and $|1| \neq 0$ by $1 \neq 0$. To verify (ii) it suffices to show that |-1| = 1. But that holds true since $|-1|^2 = |(-1)^2| = 1$ and $|-1| \ge 0$. The last claim follows by

$$-|x - y| = |x| - (|y - x| + |x|) \le |x| - |y| \le |x - y| + |y| - |y| = |x - y|$$

and

$$|x - y| = |x + (-y)| \le |x| + |-y| = |x| + |y|.$$

1.1.4 Examples (a) Obviously, the standard absolute values

$$|\cdot|_{\infty}:\mathbb{Q},\mathbb{R}\to\mathbb{R}_{\geqslant 0},\ x\mapsto\begin{cases}x\quad \text{if }x\geqslant 0\\-x\quad \text{if }x<0\end{cases}\quad\text{and}\quad |\cdot|_{\infty}:\mathbb{C}\to\mathbb{R}_{\geqslant 0},\ z\mapsto\sqrt{z\overline{z}}\end{cases}$$

are absolute values on the fields \mathbb{Q} , \mathbb{R} and \mathbb{C} , respectively. These absolute values are all archimedean since $|1+1|_{\infty} = 2 > 1$. Unless mentioned differently, we always assume \mathbb{Q} , \mathbb{R} and \mathbb{C} to be equipped with the standard absolute values. If no confusion can arise we usually write $|\cdot|$ instead of $|\cdot|_{\infty}$.

(b) The standard absolute value on the quaternions

$$|\cdot|_{\infty}: \mathbb{H} \to \mathbb{R}_{\geq 0}, \ q = a + b\,\mathbf{i} + c\,\mathbf{j} + d\,\mathbf{k} \mapsto \sqrt{\overline{q}q} = \sqrt{a^2 + b^2 + c^2 + d^2}$$

where a, b, c, d are real, is an archimedean absolute value. Usually it is briefly denoted $|\cdot|$.

(c) For every division ring R the map

$$|\cdot|: R \to \mathbb{R}, \ x \mapsto \begin{cases} 0 & \text{if } x = 0, \\ 1 & \text{else} \end{cases}$$

is a non-archimedean absolute value. It is called the *trivial absolute value* on R.

(d) An absolute value $|\cdot|: \mathbb{F} \to \mathbb{R}_{\geq 0}$ defined on a finite field \mathbb{F} has to be trivial. To see this observe that for each $x \in \mathbb{K}^{\times}$ there exists an $n \in \mathbb{N}$ such that $x^n = 1$. This entails $|x|^n = 1$, hence |x| = 1 for all $x \in \mathbb{K}^{\times}$. So $|\cdot|$ is trivial.

(e) The field of formal Laurent power series $\mathbb{K}((X))$ over a field \mathbb{K} can be equipped with an absolute value as follows. Choose $0 < \varepsilon < 1$ and define the absolute value $\left|\sum_{k \in \mathbb{Z}} a_k X^k\right|$ of an element $\sum_{n \in \mathbb{Z}} a_n X^n \in \mathbb{K}((X))$ as ε^n , where n is the minimal integer such that $a_n \neq 0$.

(f) Let p be prime number. For every integer $m \neq 0$ let $\nu_p(m)$ be the exponent of p in the prime factor decomposition of m that is $m = p^{\nu_p(n)}n$ where n is relatively prime to p. For $m \in \mathbb{Z}$ and $n \in \mathbb{N}_{>0}$ one defines the *p*-adic absolute value of the rational number $x = \frac{m}{n}$ by

$$|x|_p = \begin{cases} 0 & \text{if } m = 0 \ p^{-\nu_p(m) + \nu_p(n)} & \text{else} \ . \end{cases}$$

Note that $|x|_p$ does not depend on the particular representation of x as the quotient of integers m and n. By definition it is immediately clear that the p-adic absolute value is an absolute value on \mathbb{Q} indeed. It is non-archimedean.

1.1.5 Proposition A valued division ring $(R, |\cdot|)$ is non-archimedean if and only if the image of \mathbb{Z} under the canonical map $\mathbb{Z} \to R$ is bounded.

Proof. Assume that $(R, |\cdot|)$ is a non-archimedean valued division ring. Then, $|0 \cdot 1| = |0| = 0$ and, under the assumption that $|(n-1) \cdot 1| \leq 1$ for some $n \in \mathbb{N}_{>0}$, $|n \cdot 1| = |(n-1) \cdot 1 + 1| = \max\{|(n-1) \cdot 1|, 1\} = 1$. Hence by induction and since |-1| = 1 one obtains that $|n \cdot 1| \leq 1$ for all $n \in \mathbb{Z}$, and the image of \mathbb{Z} in R is bounded.

To show the converse assume that the image of \mathbb{Z} in R is bounded by some constant C > 0. Then, for all $x, y \in R$ and $n \in \mathbb{N}_{>0}$ by the binomial formula and the triangle inequality

$$|x+y|^{n} = \left|\sum_{k=0}^{n} \binom{n}{k} x^{k} y^{n-k}\right| \le (n+1) C \max\{|x|, |y|\}^{n}$$

Taking the *n*-th root gives $|x + y| \leq ((n + 1)C)^{1/n} \max\{|x|, |y|\}$ which after passing to the limit $n \to \infty$ entails $|x + y| \leq \max\{|x|, |y|\}$ since $\lim_{n \to \infty} ((n + 1)C)^{1/n} = 1$. Hence $(R, |\cdot|)$ is non-archimedean.

1.1.6 Proposition Let $|\cdot|$ be an absolute value on the division ring R. Then for every $\tau > 0$ with $\tau \leq 1$ the map $|\cdot|^{\tau} : R \to \mathbb{R}_{\geq 0}$ is an absolute value on R as well. It is archimedean if and only if $|\cdot|$ is archimedean.

Proof. To prove that $|\cdot|^{\tau}$ is an absolute value it suffices to show that $(a + b)^{\tau} \leq a^{\tau} + b^{\tau}$ for all $a, b \geq 0$. Without loss of generality we may assume $a \geq b > 0$. By dividing through b^{τ} one sees that the claim is equivalent to $(t+1)^{\tau} \leq t^{\tau} + 1$ for all $t \geq 1$. For t = 1 this is certainly true. The derivative of the function $h : [1, \infty) \to \mathbb{R}$, $t \mapsto (t+1)^{\tau} - t^{\tau}$ now is given by $h'(t) = \tau \left((t+1)^{\tau-1} - t^{\tau-1} \right)$ which is negative since $\tau - 1 \leq 0$ and $1 + t > t \geq 1$. Hence h is monotone decreasing and $(t+1)^{\tau} - t^{\tau} \leq 1$ for all $t \geq 1$.

Since $(0, \infty) \to \mathbb{R}$, $t \mapsto t^{\tau}$ is strictly increasing and unbounded, the image of \mathbb{Z} in R is unbounded with respect to $|\cdot|^{\tau}$.

1.1.7 An absolute value $|\cdot|: R \to \mathbb{R}_{\geq 0}$ on a division ring R induces the metric $d: R \times R \to \mathbb{R}_{\geq 0}$, $(x, y) \mapsto |x-y|$ which then gives rise to a topology on R. This topology has the following properties:

(TDR1) Addition $+ : R \times R \rightarrow R$ is continuous.

- (TDR2) Multiplication $\cdot : R \times R \rightarrow R$ is continuous.
- (TDR3) Inversion $(\cdot)^{-1} : R^{\times} \to R^{\times}$ is continuous, where R^{\times} denotes the set of units in R i.e. $R^{\times} = R \setminus \{0\}$.

Proof. Addition is continuous since for all $a, b, x, y \in R$ by the triangle inequality

$$d(x + y, a + b) = |x + y - (a + b)| \le |x - a| + |y - b| = d(x, a) + d(y, b)$$

Actually, this even shows that addition is Lipschitz continuous. Now fix $a, b \in R$ and let $C = \max\{|a|, |b|\} + 1$. Then for all $x, y \in R$ with d(y, b) < 1

$$d(x \cdot y, a \cdot b) = |(x \cdot y - a \cdot y) + (a \cdot y - a \cdot b)| \le |x - a| |y| + |a| |y - b| \le C(d(x, a) + d(y, b)).$$

Hence multiplication is continuous. Finally, fix $a \in R^{\times}$ and let $x \in R^{\times}$ with $d(x, a) < \frac{|a|}{2}$. Then $|x| \ge |a| - d(x, a) > \frac{|a|}{2} > 0$ and

$$d(x^{-1}, a^{-1}) = |x^{-1} - a^{-1}| = |x^{-1} \cdot a^{-1}| |x - a| = \frac{1}{|x||a|} d(x, a) < \frac{2}{|a|^2} d(x, a) .$$

So inversion is also continuous.

1.1.8 Definition A division ring or field R which is equipped with a topology so that (TDR1), (TDR2) and (TDR3) are satisfied is called a *topological division ring* or a *topological field*, respectively.

1.1.9 Lemma If $|\cdot|$ is a non-trivial absolute value on the division ring R, then there exists an element $t \in R^{\times}$ such that the sequence $(t^n)_{n \in \mathbb{N}}$ converges to 0. Furthermore in this case every 0-neighborhood in R contains infinitely many elements.

Proof. By non-triviality of $|\cdot|$ there exists $t \in R^{\times}$ such that $|t| \neq 1$. By possibly passing to t^{-1} we can assume |t| < 1. Since then $\lim_{n \to \infty} |t|^n = 0$, the sequence $(t^n)_{n \in \mathbb{N}}$ converges to 0. This implies in particular that for every $\varepsilon > 0$ the open ball $\mathbb{B}(0, \varepsilon) = \{t \in R \mid |t| < \varepsilon\}$ contains infinitely many elements. So the lemma is proved.

1.1.10 Definition Two absolute values $|\cdot|$ and $|\cdot|'$ on a division ring R are called *equivalent* if they induce the same topology on R.

1.1.11 Theorem Let $|\cdot|$ and $|\cdot|'$ be two absolute values on the division ring R. Then they are equivalent if and only if there exists e > 0 such that $|\cdot|' = |\cdot|^{\tau}$. In particular the trivial absolute value is the only one inducing the discrete topology on R.

Proof. Let us first show the following proposition.

(A) If $|\cdot|$ and $|\cdot|'$ are equivalent, then the relation |x| < 1 holds true for $x \in R^{\times}$ if and only if |x|' < 1.

Since $|x^{-1}| = \frac{1}{|x|}$ and $|x^{-1}|' = \frac{1}{|x|'}$ for all $x \in R^{\times}$, (A) implies that |x| > 1 if and only if |x|' > 1and that |x| = 1 if and only if |x|' = 1. To verify claim (A) assume now that 0 < |x| < 1. Then $\lim_{n \to \infty} |x^n| = 0$, hence $(x^n)_{n \in \mathbb{N}}$ converges to 0. By assumption, $\lim_{n \to \infty} |x^n|' = 0$ then holds as well which implies that |x|' < 1. By switching $|\cdot|$ and $|\cdot|'$ the converse holds true, so (A) is proved.

Next we show that $|\cdot|$ is trivial if and only if the induced topology on R is discrete. Namely, if $|\cdot|$ is non-trivial, then there exists $x \in R^{\times}$ such that $|x| \neq 1$. After possibly passing to $\frac{1}{x}$ we can achieve that |x| < 1. So $\lim_{n \to \infty} |x^n| = 0$, which means that $(x^n)_{n \in \mathbb{N}}$ is a sequence of non-zero elements of R converging to 0. But this implies that the singleton $\{0\}$ is not open in the topology induced by $|\cdot|$, hence this topology is non-discrete. Since obviously the trivial absolute value induces the discrete topology on R the second claim of the theorem is proved.

Now assume that $|\cdot|' = |\cdot|^{\tau}$ for some $\tau > 0$. Then a subset $B \subset R$ is a metric open ball with respect to $|\cdot|$ if and only if it is one with respect to $|\cdot|'$ since for $x \in R$ and $\varepsilon > 0$

$$\left\{ y \in R \mid |y - x| < \varepsilon \right\} = \left\{ y \in R \mid |y - x|' < \varepsilon^{\tau} \right\} \text{ and }$$
$$\left\{ y \in R \mid |y - x|' < \varepsilon \right\} = \left\{ y \in R \mid |y - x| < \varepsilon^{1/\tau} \right\}.$$

Hence the open sets with respect to the metric defined by $|\cdot|$ coincide with those defined by $|\cdot|'$ and the two absolute values are equivalent.

Let us finally show the other direction and assume that $|\cdot|$ and $|\cdot|'$ are equivalent. By the already proven second claim of the theorem we can restrict to the case where the induced topology is non-discrete which means to the case where both $|\cdot|$ and $|\cdot|'$ are non-trivial. We show that there exists

 $\tau > 0$ such that $|x|' = |x|^{\tau}$ for all $x \in R^{\times}$ with |x| > 1. This is sufficient, since if |x| = 1, then $|x|' = 1 = |x|^{\sigma}$ for any $\sigma > 0$ by (A), and since if $x \in R^{\times}$ with |x| < 1 then $|x^{-1}| > 1$ and

$$|x|' = \frac{1}{|x^{-1}|'} = \frac{1}{|x^{-1}|^{\tau}} = |x|^{\tau}$$

The existence of a $\tau > 0$ with the claimed property is equivalent to the function

$$R^{\times} \to \mathbb{R}, \ x \mapsto \frac{\ln |x|'}{\ln |x|}$$

being constant. Assume that that is not the case. Then there exist $x, y \in \mathbb{R}^{\times}$ with |x|, |y| > 1 such that $\frac{\ln |x|'}{\ln |x|} \neq \frac{\ln |y|'}{\ln |y|}$. By possibly switching x and y we can assume $\frac{\ln |x|'}{\ln |x|} < \frac{\ln |y|'}{\ln |y|}$. But that implies $\frac{\ln |x|'}{\ln |y|'} < \frac{\ln |x|}{\ln |y|} < \frac{\ln |x|}{\ln |y|}$ since the logarithms are positive by assumptions on x and y and (A). Hence there exists a rational number $\frac{p}{q}$ with $p, q \in \mathbb{N}_{>0}$ such that

$$\frac{\ln |x|'}{\ln |y|'} < \frac{p}{q} < \frac{\ln |x|}{\ln |y|} \; .$$

Then $|x^q|' < |y^p|'$ and $|y^p| < |x^q|$ which entails

$$\left| rac{x^q}{y^p}
ight|' < 1 \, \, {
m and} \, \, \left| rac{x^q}{y^p}
ight| > 1 \, \, .$$

This contradicts (A) and the theorem is proved.

1.1.12 Remarks (a) By Ostrowski's theorem (Ostrowski, 1916, p. 276), see also (Gouvêa, 1997, Thm. 3.1.3), every non-trivial absolute value on the field \mathbb{Q} of rational numbers is either equivalent to the standard absolute value $|\cdot|_{\infty}$ or to a *p*-adic absolute value $|\cdot|_p$ for some prime number *p*. Observe that for different primes *p* and *q* the absolute values $|\cdot|_p$ and $|\cdot|_q$ are not equivalent.

(b) Another theorem of Ostrowski (Ostrowski, 1916, p. 284), sometimes called big Ostrowski's theorem, tells that for every archimedean valued field $(\mathbb{K}, |\cdot|)$ there exists an embedding $\iota : \mathbb{K} \hookrightarrow \mathbb{C}$ into the field of complex numbers with its standard absolute value and a positive real number $\tau \leq 1$ such that

$$|x| = |\iota(x)|_{\infty}^{\tau}$$
 for all $x \in \mathbb{K}$.

In particular this means that every complete archimedean valued field is isomorphic to either $(\mathbb{R}, |\cdot|_{\infty}^{\tau})$ or $(\mathbb{C}, |\cdot|_{\infty}^{\tau})$ for some positive $\tau \leq 1$.

(c) The *p*-adic absolute values on \mathbb{Q} have extensions to \mathbb{R} by (Lang, 2002, XII, §4, Thm. 4.1). This is a highly non-obvious result. To prove it one has to check first that $|\cdot|_p$ can be extended to an absolute value $|\cdot|$ on the field k of real numbers algebraic over \mathbb{Q} . This extended absolute value is, and that turns out to be crucial, again non-archimedean. Now one observes that $|\cdot|$ can be extended to the polynomial ring $\mathbb{K}[X]$ by the *Gauß norm* $|p(X)| = \max_{0 \le i \le n} \{a_i\}$ where $p(X) = a_n X^n + \ldots + a_1 X + a_0 \in \mathbb{K}[X]$. The Gauß norm obviously extends to an absolute value on the fraction field $\mathbb{k}(X)$. Again, this extension is non-archimedean. Now one recalls that \mathbb{R} is a purely transcendental field extension of k and uses a transfinite induction type argument involving the just constructed Gauß norm to extend $|\cdot|$ from K to R. The thus obtained extension of the *p*-adic absolute value to \mathbb{R} is not unique. In its construction, the axiom of choice is used, so one can not even give an explicit formula for such an extension.

A.1.2. The category of topological vector spaces

Vector space topologies

1.2.1 Definition Let R be a topological division ring. A topology T on a vector space E over R is called a *vector space topology* if the following axioms hold true:

(TVS1) Addition $+ : E \times E \rightarrow E$ is continuous.

(TVS2) Multiplication by scalars $\cdot : R \times E \rightarrow E$ is continuous.

The topology \mathcal{T} on E is called *translation invariant* if for every $w \in E$ the linear map $\ell_w : E \to E$, $v \mapsto v + w$ is a homeomorphism.

A vector space E endowed with a vector space topology on it is called a *topological vector space* (*over* R), for short a tvs

1.2.2 Remark Let us recall at this point some notation from linear algebra. Assume that V is a left vector space over the divison ring R. If $A, B \subset V$ are two non-empty subsets, then A + B is the set of all $v \in V$ for which there exist $x \in A$ and $y \in B$ such that v = x + y. If A or B is empty, then A + B is defined as the empty set. In case A is a singleton that is if $A = \{x\}$, then we often write x + B instead of $\{x\} + B$. If $B \subset \mathcal{P}(V)$ is a non-empty set of subsets of V, then we denote by A + B and x + B the sets $\{A + B \in \mathcal{P}(V) \mid B \in B\}$ and $\{x + B \in \mathcal{P}(V) \mid B \in B\}$, respectively. If $A \subset \mathcal{P}(V)$ is a second non-empty set of subsets of V, then A + B stands for the set of all sets of the form A + B, where $A \in A$ and $B \in B$.

In case C is a subset of the ground ring R, then $C \cdot A$ is defined as the set of all $v \in V$ for which there exist $r \in C$ and $x \in A$ such that $v = r \cdot x$. If $r \in R$ we write $r \cdot A$ for $\{r\} \cdot A$. Likewise, if $x \in V$, $C \cdot x$ stands for $C \cdot \{x\}$. Analogously as for addition the sets $\mathcal{C} \cdot A$, $C \cdot \mathcal{A}$ and $\mathcal{C} \cdot \mathcal{A}$ are defined when $\mathcal{C} \subset \mathcal{P}(R)$ and $\mathcal{A} \subset \mathcal{P}(V)$ are non-empty.

1.2.3 Proposition Let E be a tvs over a topological division ring R. Then the following holds true:

- (i) For every $r \in R^{\times}$ and $w \in E$ the homothety $\ell_{r,w} : E \to E$, $v \mapsto rv + w$ is a homeomorphism with inverse $\ell_{r^{-1},-r^{-1}w}$.
- (ii) Let w be an element of E and $r \in R^{\times}$. A filter base \mathcal{B} on E then is a filter base for the zero neighborhoods if and only if $w + r\mathcal{B}$ is a filter base for the neighborhoods of w.
- (iii) If \mathcal{B} is a filter base of the filter of zero neighborhoods, then the closure of any non-empty $A \subset E$ is given by

$$\bar{A} = \bigcap_{U \in \mathcal{B}} A + U \; .$$

- (iv) Let $A \subset E$ be open and $B \subset E$. Then the set A + B is open.
- (v) Let $A, B \subset E$ be closed and assume that A is quasi-compact that is that any filter on A has a cluster point. Then the set A + B is closed.
- (vi) The space E is ?? or, equivalently, each point of E possesses a neighborhood base consisting of closed subsets.

Proof. ad (i). The homothety $\ell_{r,w}$ is continuous since addition and multiplication by a scalar are continuous maps on a tvs Since for all $v \in V$

$$\begin{split} \ell_{r^{-1},-r^{-1}w} \circ \ell_{r,w}(v) &= r^{-1}(rv+w) - r^{-1}w = v, \text{ and} \\ \ell_{r,w} \circ \ell_{r^{-1},-r^{-1}w}(v) &= r(r^{-1}v - r^{-1}w) + w = v \end{split}$$

the homothety $\ell_{r,w}$ is invertible, and its inverse is $\ell_{r^{-1},-r^{-1}w}$.

ad (ii). This follows since $\ell_{r,w}$ is a homeomorphism.

ad (iii). Let $B = \bigcap_{U \in \mathcal{B}} A + U$. Let v be an element of the closure of A. Then, for $U \in \mathcal{B}$, there exists an element $a \in A \cap v - U$ by (ii) and since -U is a zero neighborhood. Hence $v \in a + U$, and $\overline{A} \subset B$ follows. Now let $v \in B$ and V be a neighborhood of v. Then there exists $U \in \mathcal{B}$ such that $v - U \subset V$. By definition of B there exists an element $a \in A$ such that $v \in a + U$. Hence $a \in v - U \subset V$ which implies that $v \in \overline{A}$. So $B \subset \overline{A}$.

ad (iv). The set A + B is either empty or coincides with the union $\bigcup_{v \in B} v + A$. In the latter case, each of the sets v + A is non-empty and open by continuity of addition. So A + B is open under the assumptions made.

ad (v). We can assume that A and B are non-empty because the claim is trivial otherwise. Assume that A + B is not closed. Then there exists an element $v \in E \setminus (A + B)$ such that each neighborhood of v meets A + B. This means in particular that the restriction of the neighborhood filter \mathcal{U} of v to A + B is a filter base. Consequently, $(-B + \mathcal{U}) \cap A$ is a filter base on A, hence possesses an accummulation point $x \in A$. For each neighborhood $V \in \mathcal{U}$ the point x is then contained in the closure of -B + V. Hence, by (iii), x is contained in v - B + U + U for every zero neighborhood U. Since by continuity of addition U + U runs through a base of zero neighborhoods when U runs through the zero neighborhoods, $x \in v - \overline{B} = v - B$ follows. Since $x \in A$ this contradicts the assumption $v \in A + B$ and A + B has to be closed.

ad (vi). Let $v \in E$, $A \subset E$ closed, and assume $v \notin A$. Choose an open neighborhood V of v such that $V \cap A = \emptyset$. Then there exists an open zero neighborhood U such that $v + U + U \subset V$. By possibly passing to $U \cap (-U)$ we can assume that U = -U. Now v + U is an open neighborhood of v and A + U one of A. These neighborhoods are disjoint because if the intersection $v + U \cap A + U$ is non-empty, then there exists an element $w \in v + U + U \cap A$ since -U = U. This contradicts $V \cap A = \emptyset$, so v + U and A + U are disjoint neighborhoods of v and A, respectively. Hence E satisfies ??.

1.2.4 Corollary Every vector space topology on a vector space over a topological division ring is translation invariant.

Proof. This follows immediately by Proposition 1.2.3 (i).

1.2.5 Definition A subset C of a vector space E over a valued division ring $(R, |\cdot|)$ is called

- (i) symmetric if $-v \in C$ for all $v \in C$,
- (ii) circled or balanced if $rv \in C$ for all $v \in C$ and $r \in R$ with $|r| \leq 1$.

1.2.6 Remark Symmetry of a subset of a vector space of a division ring is even defined when the underlying division ring does not carry an absolute value.

1.2.7 Lemma Let C be a subset of a topological vector space E over a valued division ring $(R, |\cdot|)$ and $r \in R$.

- (i) If C is symmetric, then the closure \overline{C} and the interior \mathring{C} are symmetric.
- (ii) If C is circled, then the closure \overline{C} and the union $\mathring{C} \cup \{0\}$ are circled.
- (iii) The set rC is symmetric (respectively circled) if C has that property.

Proof. Without loss of generality we can assume $C \neq \emptyset$. Claim (i) then follows immediately since multiplication by -1 is a homeomorphism. To prove claim (ii) assume that C is circled. Let $s \in R$ with $|s| \leq 1$. Assume $v \in \overline{C}$ and consider sv. We have to show that $sv \in \overline{C}$. If s = 0 then $sv = 0 \in C \subset \overline{C}$ since C is circled. So we can assume $s \neq 0$ and need to show that for every neighborhood V of sv the intersection $C \cap V$ is non-empty. Since |s| > 0, the homothety $\ell_s : E \to E$, $w \mapsto sw$ is a homeomorphism with inverse $\ell_{s^{-1}}$. Hence $s^{-1}V$ is a neighborhood of v. Since v lies in the closure of C there exists an element $w \in C \cap s^{-1}V$. Hence $sw \in C \cap V$ by assumption on C and \overline{C} is circled.

If $v \in \mathring{C} \cup \{0\}$ then $0 = 0 \cdot v \in \mathring{C} \cup \{0\}$. It remains to show that $sv \in \mathring{C} \cup \{0\}$ for $s \in R$ with $0 < |s| \le 1$ and $v \in \mathring{C} \setminus \{0\}$. Under this assumption the homothety ℓ_s is a homeomorphism, so $s\mathring{C}$ is an open subset of C since C is circled. Hence $sv \in s\mathring{C} \subset \mathring{C}$, and $\mathring{C} \cup \{0\}$ is circled as well.

Claim (iii) follows immediately from the observation that for $v \in C$ and $s \in R$ the relation $srv \in rC$ holds true if $sv \in C$.

1.2.8 Proposition and Definition The intersection of a non-empty family $(C_i)_{i \in I}$ of symmetric (respectively circled) subsets $C_i \subset E$, $i \in I$ of a topological vector space E over a valued division ring $(R, |\cdot|)$ is symmetric (respectively circled). In particular, if $A \subset E$ is a subset, then the sets

$$\operatorname{Sym} A = \bigcap_{\substack{A \subset B \subset E \\ B \text{ is symmetric}}} B \text{ and } \operatorname{Circ} A = \bigcap_{\substack{A \subset B \subset E \\ B \text{ is circled}}} B$$

are symmetric and circled, respectively. They have the property that Sym A is the smallest symmetric and Circ A the smallest circled subsets of E containing A. They are called the symmetric and the circled hull of A, respectively. Analogously,

$$\overline{\operatorname{Sym}} A = \bigcap_{\substack{A \subset B = \bar{B} \subset E \\ B \text{ is symmetric}}} B \text{ and } \overline{\operatorname{Circ}} A = \bigcap_{\substack{A \subset B = \bar{B} \subset E \\ B \text{ is circled}}} B$$

are called the closed symmetric and the closed circled hull of A, respectively. They have the property that $\overline{\text{Sym}} A$ is the smallest closed symmetric and $\overline{\text{Circ}} A$ the smallest closed circled subset of E containing A.

Proof. Note first that all the hulls in the proposition are well-defined since E is closed and circled. Let C denote the intersection of the family $(C_i)_{i \in I}$. Assume that for some $r \in R$ with $|r| \leq 1$ the inclusion $rC_i \subset C$ holds true for all $i \in I$. Then $rC \subset C$, hence if all C_i are symmetric (respectively circled), so is C. This observation now entails that $\operatorname{Sym} A$ is symmetric, Circ is circled, $\overline{\operatorname{Sym}} A$ is closed and symmetric, and finally that $\overline{\operatorname{Circ}} A$ is closed and circled. Moreover, all those sets contain A. The minimality properties of these sets are clear by construction. **1.2.9 Remark** Observe that by the proposition A is symmetric if and only if Sym A = A and circled if and only if Circ A = A. Analogously, $\overline{Sym} A = A$ if and only if A is closed symmetric and $\overline{Circ} A = A$ if and only if A is closed and circled.

1.2.10 Lemma Let E be a topological vector space over the valued division ring $(R, |\cdot|)$ and $A \subset E$ non-empty. Then

Sym
$$A = A \cup -A$$
 and Circ $A = \bigcup_{r \in R, |r| \leq 1} rA$.

For the closed hulls one has

 $\overline{\operatorname{Sym}} A = \overline{\operatorname{Sym} A} \quad \text{and} \quad \overline{\operatorname{Circ}} A = \overline{\operatorname{Circ} A} \;.$

Proof. Since $A \cup -A$ is symmetric by definition, contains A, and is contained in Sym A, the equality Sym $A = A \cup -A$ holds true. Similarly, $\bigcup_{r \in R, |r| \leq 1} rA$ is circled by definition, contains A, and is contained in Circ A by definition of the circled hull. Hence Circ $A = \bigcup_{r \in R, |r| \leq 1} rA$. The remainder of the claim follows from Lemma 1.2.7.

1.2.11 Definition Assume that B, C are subsets of a vector space E over the valued division ring $(R, |\cdot|)$. Then one says that

- (i) C absorbes B if there exists a real number $t \in \mathbb{R}_{\geq 0}$ such that $B \subset rC$ for all $r \in R$ with $|r| \geq t$,
- (ii) C is absorbing or absorbent if C absorbes every one-point set of E that is if for every $v \in E$ there exists $t \in \mathbb{R}_{\geq 0}$ such that $v \in rC$ for all $r \in R$ with $|r| \geq t$.

If the vector space E carries in addition a vector space topology, then one says that

(iii) the subset $B \subset E$ is *bounded* if it is absorbed by every zero neighborhood.

1.2.12 Lemma Let E be a vector space over the valued division ring $(R, |\cdot|)$. Then the following holds true.

- (i) If C_1, \ldots, C_n are absorbing subset of E, then the intersection $C_1 \cap \ldots \cap C_n$ is absorbing.
- (ii) If C is an absorbing subset of E, then rC is absorbing for every $r \in R^{\times}$.

Proof. ad (i). Let $v \in E$ and choose $t_1, \ldots, t_n \in \mathbb{R}_{\geq 0}$ such that $v \in rC_i$ for $|r| \geq t_i$. Put $t = \max\{t_1, \ldots, t_n\}$. Then $v \in r(C_1 \cap \ldots \cap C_n)$ for $|r| \geq t$, hence $C_1 \cap \ldots \cap C_n$ is absorbing.

ad (ii). Choose $t \in \mathbb{R}_{\geq 0}$ such that $v \in sC$ for all $s \in R$ with $|s| \geq t$. Then one has $|sr| \geq t$ for all $s \in R$ with $|s| \geq \frac{t}{|r|}$, hence $v \in s(rC)$ for all such s. Therefore rC is absorbing.

1.2.13 Proposition The filter of zero neighborhoods of a topological vector space E over $(R, |\cdot|)$ has a filter base \mathcal{B} with the following properties:

- (i) For each $V \in \mathcal{B}$ there exists $U \in \mathcal{B}$ such that $U + U \subset V$.
- (ii) Every element $V \in \mathcal{B}$ is circled and absorbing.
- (iii) There exists an element $r \in R^{\times}$ with 0 < |r| < 1 such that $V \in \mathcal{B}$ implies $rV \in \mathcal{B}$.

Conversely, if \mathcal{B} is a filter base on an *R*-vector space E such that (i) to (iii) hold true, then there exists a unique vector space topology on E such that \mathcal{B} is a neighborhood base at the origin. In case the ground ring *R* is archimedean, a filter base on E which satisfies (i) and (ii) already induces a unique vector space topology having \mathcal{B} as a neighborhood base at 0. In either of these two cases, the thus constructed topology coincides with the coarsest translation invariant topology for which \mathcal{B} is a set of zero neighborhoods.

Proof. Assume that E is a tvs Let \mathcal{B} be the set of circled neighborhoods of 0. We show first that \mathcal{B} is a base of the filter \mathcal{U}_0 of zero neighborhoods. Let $W \in \mathcal{U}_0$. By Axiom (TVS2) there exists an $\varepsilon > 0$ and an open zero neighborhood U such that $sU \subset W$ for all $s \in R$ with $|s| < \varepsilon$. Then $V = \bigcup_{s \in R^{\times} \& |s| < \varepsilon} sU$ is a zero neighborhood since by Lemma 1.1.9 the set of $s \in R^{\times}$ with $|s| < \varepsilon$ is

non-empty. By construction V is contained in W and circled, so $V \in \mathcal{B}$. Hence \mathcal{B} is a filter base of \mathcal{U}_0 .

Next recall that there exists $r \in R^{\times}$ with 0 < |r| < 1 since the absolute value $|\cdot|$ is non-trivial. Let $V \in \mathcal{B}$. Then $sV \subset V$ for all $s \in R$ with $|s| \leq 1$ which entails $srV \subset rV$ for all such s. Hence rV is circled and an element of \mathcal{B} as well. This proves (iii). Since addition on E is continuous, there exist for given $V \in \mathcal{B}$ open neighborhoods U_1, U_2 of the origin such that $U_1 + U_2 \subset V$. Choose $U \in \mathcal{B}$ such that $U \subset U_1 \cap U_2$. Then $U + U \subset V$ and (i) is proved. To show that any $V \in \mathcal{B}$ is absorbing let $v \in E$. By continuity of scalar multiplication there exists $\varepsilon > 0$ such that $sv \in V$ for all $s \in R$ with $|s| < \varepsilon$. By Proposition 1.2.3 (i) this entails $v \in sV$ for all $s \in R$ with $|s| > \varepsilon$ and V is absorbing.

Now assume that E is an R-vector space and that \mathcal{B} is a filter base that satisfies (i), (ii) and, if $|\cdot|$ is non-archimedean, (iii). Since \mathcal{B} consists of non-empty circled sets, $0 \in V$ for all $V \in \mathcal{B}$. Let $\mathfrak{T} \subset \mathfrak{P}(\mathcal{E})$ be the set of all $U \subset \mathcal{E}$ such that for each $v \in U$ there exists $V \in \mathfrak{B}$ with $v + V \subset U$. By definition and since \mathcal{B} is a filter base, \mathcal{T} is a topology on E. By construction, \mathcal{T} is also the coarsest translation invariant topology for which $\mathcal B$ is a set of zero neighborhoods. We show that $\mathcal B$ is a base of the filter \mathcal{U}_0 of zero neighborhoods. By definition of \mathcal{T} there exists for each $U \in \mathcal{U}_0$ a $V \in \mathcal{B}$ such that $V \subset U$. So it remains to show that each $V \in \mathcal{B}$ is a zero neighborhood. To this end let U be the set of all $v \in V$ for which there exists a $W \in \mathcal{B}$ with $v + W \subset V$. Since $0 + V \subset V$ one has $0 \in U$. The relation $U \subset V$ holds because $0 \in W$ for all $W \in \mathcal{B}$. Now let $v \in U$. By (i) there exists W' such that $v + W' + W' \subset V$ which entails $v + W' \subset U$. Hence $U \in \mathfrak{T}$ and V is a zero neighborhood. Next we verify that \mathcal{T} is a vector space topology. We start with continuity of addition. Let W be an open neighborhood of v + w, where $v, w \in E$. Then there exists $V \in \mathcal{B}$ such that $v + w + V \subset W$. Choose $U \in \mathcal{B}$ such that $U + U \subset V$. Then v + U and w + U are neighborhoods of v and w, respectively, and $(v+U) + (w+U) \subset v + w + V \subset W$. So addition is continuous. We continue with scalar multiplication. Let W be an open neighborhood of rv, where $r \in R$ and $v \in E$. Then there exists $V \in \mathcal{B}$ such that $rv + V + V \subset W$. Since V is absorbing by (ii) there exists $\varepsilon > 0$ such that $(s-r)v \in V$ for all $s \in R$ with $|s-r| < \varepsilon$. Now if $|\cdot|$ is non-archimedean choose $t \in R^{\times}$ according to (iii), and put $V_n = t^n V$ for all $n \in \mathbb{N}$. In the archimedean case let $t = \frac{1}{2}$ and use (i) to construct recursively a sequence $(V_n)_{n\in\mathbb{N}}$ of elements of \mathcal{B} such that $2^nV_n = V_n + \cdots + V_n \subset V$, where the sum has 2^n summands. In either of these cases, choose $N \in \mathbb{N}$ large enough so that $|t|^N < \frac{1}{|r|+\varepsilon}$. Then $V_N \in \mathcal{B}$ and $v + V_N$ is a neighborhood of v. Moreover, for $w \in v + V_N$ there exists an element $x \in V$ such that $w - v = t^N x$. Then the relation $s(w - v) = st^N x \in V$ holds whenever $|s - r| < \varepsilon$ since V_N is circled. Hence for such w and s

$$sw = rv + s(w - v) + (s - r)v \in rv + V + V \subset W$$

This means that scalar multiplication is continuous, and the proof is finished.

Morphisms of topological vector spaces

1.2.14 Definition By a *morphism* of topological vector spaces over the topological division ring R one understands a continuous R-linear map $f : E \to F$ between two topological vector spaces E and F over R. The space of morphisms between E and F will be denoted $\operatorname{Hom}_{R-\mathsf{TVS}}(E, F)$ or just $\operatorname{Hom}_R(E, F)$ or $\operatorname{Hom}(E, F)$ if now confusion can arrise.

1.2.15 Theorem The topological vector spaces over a topological division ring R as objects together with their morphisms form an additive category which we denote by R-TVS. More precisely, R-TVS is a category enriched over the category of R-vector spaces where addition and scalar multiplication on the hom-spaces Hom(E, F) are given by

 $\begin{aligned} +: \operatorname{Hom}(\mathbf{E}, \mathbf{F}) \times \operatorname{Hom}(\mathbf{E}, \mathbf{F}) &\to \operatorname{Hom}(\mathbf{E}, \mathbf{F}), \ (f, g) \mapsto f + g = (\mathbf{E} \ni v \mapsto f(v) + g(v) \in \mathbf{F}) \\ \cdot: R \times \operatorname{Hom}(\mathbf{E}, \mathbf{F}) \to \operatorname{Hom}(\mathbf{E}, \mathbf{F}), \ (r, f) \mapsto r \cdot f = (\mathbf{E} \ni v \mapsto r \cdot f(v) \in \mathbf{F}) \end{aligned}$

Proof. Observe first that the identity map id_E on a topological vector space E is linear and continuous and so is the composition $g \circ f$ of two morphisms of topological vector spaces $f : E \to F$ and $g : F \to G$. Hence topological vector spaces over R together with linear and continuous maps between them form a category.

Next check that the hom-space $\operatorname{Hom}(E, F)$ is an abelian group. Associativity and commutativity of addition follow from the respective properties on F. The zero element is the constant map $E \to F$, $v \mapsto 0$ and the inverse of a morphism $f : E \to F$ is given by $-f : E \to F$, $v \mapsto -f(v)$. Similarly one checks that multiplication by scalars on $\operatorname{Hom}(E, F)$ is associative and distributes from the left and from the right over addition since scalar multiplication on F has these properties. Finally, the unit of R acts as identity on $\operatorname{Hom}(E, F)$ since it does so on F. Hence $\operatorname{Hom}(E, F)$ carries the structure of an R left vector space.

Composition of morphisms $Hom(E, F) \times Hom(F, G) \rightarrow Hom(E, G)$, $(f, g) \rightarrow g \circ f$ is an *R*-bilinear map as the following equalities for $f, f_1, f_2 \in Hom(E, F)$, $g, g_1, g_2 \in Hom(F, G)$, $r \in R$, and $v \in E$ show:

$$(f \circ (g_1 + g_2))(v) = f((g_1 + g_2)(v)) = f(g_1(v) + g_2(v)) = = f \circ g_1(v) + f \circ g_2(v) = (f \circ g_1 + f \circ g_2)(v) , (f \circ (rg))(v) = f((rg)(v)) = f(rg(v)) = rf(g(v)) = (r(f \circ g))(v) , ((f_1 + f_2) \circ g)(v) = (f_1 + f_2)(g(v)) = f_1(g(v)) + f_2(g(v)) = = f_1 \circ g(v) + f_2 \circ g(v) = (f_1 \circ g + f_2 \circ g)(v) , ((rf) \circ g)(v) = (rf)(g(v)) = r(f(g(v))) = r(f \circ g(v)) = (r(f \circ g))(v) .$$

Hence R-TVS is a category enriched over the category of R-vector spaces. In particular, R-TVS then is an additive category.

1.2.16 Example For every tvs E and non-zero element t of the ground ring R the map $\ell_t : E \to E$, $v \mapsto tv$ is an isomorphism of topological vector spaces by Proposition 1.2.3 (i).

1.2.17 Proposition and Definition A linear map $f : E \to F$ between topological vector spaces over a valued division ring $(R, |\cdot|)$ maps symmetric sets to symmetric sets and circled sets to circled sets. If in addition f is continuous, then f is bounded that means it maps bounded subsets of E to bounded subsets of F.

Proof. Since by linearity f(tv) = tf(v) for all $v \in E$ and $t \in R$, f(C) is symmetric (respectively circled) if the subset $C \subset E$ is.

To verify the second claim let $B \subset E$ be bounded and $V \subset F$ a zero neighborhood. Then $f^{-1}(V)$ is a zero neighborhood in E by continuity of f. Hence there exists an $r \in \mathbb{R}_{\geq 0}$ such that $B \subset tf^{-1}(V)$ for all $t \in R$ with $|t| \geq r$. By linearity of f one obtains $f(B) \subset tV$ for all such t, so f is bounded. \Box

1.2.18 Remark By the proposition continuity of a linear map between topological vector spaces implies the map to be bounded. As we will see later in this monograph, the converse does in general not hold true unless the underlying topological vector spaces are for example normable.

Normed real division algebras and local convexity

1.2.19 The major class of topological divison rings over which topological vector spaces are defined is formed by valued division rings $(R, |\cdot|)$ which carry the structure of an \mathbb{R} -algebra such that for all $r \in \mathbb{R}$ and $x \in R$ the equality

$$|rx| = |r|_{\infty} \cdot |x|$$

holds true. We will therefore given them a particular name and call them *normed real division algebras*. Note that the field of real numbers can be embedded into a normed real division algebra R by the natural map $\mathbb{R} \mapsto R$, $r \mapsto r \cdot 1$. Since \mathbb{R} with its standard absolute value is archimedean, so is every normed real division algebra. By the Frobenius theorem, Frobenius (1878), there exist only three finite dimensional real division algebras, namely the field of real numbers \mathbb{R} , the field of complex numbers \mathbb{C} , and the quaternions \mathbb{H} .

1.2.20 Definition Under the assumption that R is a normed real division algebra one calls a subset $C \subset E$ of an R-vector space

(i) convex if $tv + (1-t)w \in C$ for all $v, w \in C$ and $t \in \mathbb{R}$ with $0 \leq t \leq 1$,

(ii) absolutely convex if $rv + sw \in C$ for all $v, w \in C$ and $r, s \in R$ such that $|r| + |s| \leq 1$,

(iii) a *cone* if $tv \in C$ for all $v \in C$ and $t \in \mathbb{R}$ with $0 \leq t \leq 1$.

1.2.21 Lemma Let R be a normed real division algebra. A subset C of an R-vector space E then is absolutely convex if and only if it is circled and convex.

Proof. The claim is trivial when $C = \emptyset$, so we assume that C is nonempty.

Let C be absolutely convex. Since C contains at least one element v one has $0 = 0 \cdot v + 0 \cdot v \in C$. Hence $rv = (1 - |r|) \cdot 0 + rv \in C$ for all $v \in C$ and $r \in R$ with $|r| \leq 1$. So C is circled. By definition of absolute convexity C is convex.

If C is circled and convex, then it contains with elements v, w also rv + sw if $|r| + |s| \le 1$. To see this observe first that $\varrho v \in C$ and $\sigma w \in C$ where the elements $\varrho, \sigma \in R$ have been chosen so that

 $|\varrho| = |\sigma| = 1$, $r = |r| \cdot \varrho$ and $s = |s| \cdot \sigma$. Now if |r| + |s| = 0, then $rv + sw = 0 \in C$ since C is circled. If |r| + |s| > 0, then

$$rv + sw = (|r| + |s|) \left(\frac{|r|}{|r| + |s|} \varrho v + \frac{|s|}{|r| + |s|} \sigma w \right) \in C$$

since C is convex and circled. Hence C is absolutely convex.

1.2.22 Lemma A linear map $f : E \to F$ between vector spaces over a normed real divison algebra R maps convex sets to convex sets, absolutely convex sets to absolutely convex sets, and cones to cones.

Proof. This an immediate consequence of the linearity of f.

1.2.23 Lemma Let E be a tvs over a normed real division algebra R, let $C, D \subset E$ be convex and $r \in R$. Then the following holds true.

- (i) The closure \overline{C} and the interior \mathring{C} are convex.
- (ii) The sets C + D and rC are convex.
- (iii) If C is absolutely convex, then so are \overline{C} and C.
- (iv) If C is absolutely convex, then so is rC for each $r \in R^{\times}$.

Proof. We consider only the cases $C, D \neq \emptyset$ because otherwise the claim is trivial.

ad (i). Let $t \in (0, 1)$. Then $t\overline{C} + (1-t)\overline{C} \subset \overline{C}$ by continuity of the map $E \times E \to E$, $(v, w) \mapsto tv + (1-t)w$. Hence \overline{C} is convex. Now let v, w be points of the interior of C and z = tv + (1-t)w. Then $z \in C$, and there exists a zero neighborhood U such that $v + U \subset C$ and $w + U \subset C$. Let $u \in U$ and compute

$$z + u = tv + (1 - t)w + tu + (1 - t)u = t(v + u) + (1 - t)(w + u) .$$

Since both v + u and w + u are elements of C so is z + u by convexity of C. Hence $z + U \subset C$ and z lies in the interior of C.

ad (ii). If $v, w \in C$, $x, y \in D$ and $t \in (0, 1)$, then by convexity of C and D

$$t(v+x) + (1-t)(w+y) = (tv + (1-t)w) + (tx + (1-t)y) \in C + D.$$

Hence C + D is convex. Similarly,

$$t(rv) + (1-t)(rw) = r(tv + (1-t)w) \in rC ,$$

so rC is convex as well.

ad (iii). Let C be absolutely convex. If $\mathring{C} \neq \emptyset$, then $0 \in \frac{1}{2}\mathring{C} - \frac{1}{2}\mathring{C} \subset C$, hence $0 \in \mathring{C}$. By Lemma 1.2.7 and (i) the claim now follows.

ad (iv). By (ii), rC is convex, so it remains to show that rC is circled. Assume that $v \in rC$. Then v = rw for a unique $w \in C$. Since C is circled, $tw \in C$ for every $t \in R$ with $|t| \leq 1$. Hence $tv = r(tw) \in rC$ for such t and rC is circled.

1.2.24 Proposition and Definition The intersection of a non-empty family $(C_i)_{i \in I}$ of convex (respectively absolutely convex) subsets $C_i \subset E$, $i \in I$ of a topological vector space E over a normed real division algebra R is convex (respectively absolutely convex). In particular, if $A \subset E$ is a subset, then the sets

$$\operatorname{Conv} A = \bigcap_{\substack{A \subset B \subset E \\ B \text{ is convex}}} B \quad \text{and} \quad \operatorname{AConv} A = \bigcap_{\substack{A \subset B \subset E \\ B \text{ is absolutely convex}}} B$$

are convex and absolutely convex, respectively. The set Conv A is called the convex hull of A and is the smallest convex set containing A. Similarly, AConv A is the smallest absolutely convex set containing A. It is called the absolutely convex hull of A. The closed convex hull $\overline{Conv} A$ and the closed absolutely convex hull $\overline{AConv} A$ of A are defined by

$$\overline{\operatorname{Conv}} A = \bigcap_{\substack{A \subset B = \bar{B} \subset E \\ B \text{ is convex}}} B \quad \text{and} \quad \overline{\operatorname{AConv}} A = \bigcap_{\substack{A \subset B = \bar{B} \subset E \\ B \text{ is absolutely convex}}} B$$

These sets have the property that $\overline{\text{Conv}} A$ is the smallest closed convex subset and $\overline{\text{AConv}} A$ the smallest closed absolutely convex subset of E containing A.

Proof. Let C be the intersection $\bigcap_{i \in I} C_i$ and assume that each C_i is absolutely convex. Let $v, w \in C$ and $r, s \in R$ with $|r| + |s| \leq 1$. Then $v, w \in C_i$, hence $rv + sw \in C_i$ for all $i \in I$. Therefore $rv + sw \in C$ and C is absolutely convex. This argument also shows that C is convex if all C_i are convex. The rest of the claim follows as in the proof of Proposition and Definition 1.2.8.

1.2.25 Remark The proposition in particular entails that A is convex if and only if Conv A = A and absolutely convex if and only if AConv A = A. Analogously, $\overline{Conv} A = A$ if and only if A is closed and convex, and $\overline{AConv} A = A$ if and only if A is closed and absolutely convex.

1.2.26 Lemma Let $A \subset E$ be a non-empty subset of a tvs E over a normed real division algebra R. Then

Conv
$$A = \left\{ \sum_{i=1}^{k} t_i v_i \in E \mid k \in \mathbb{N}_{>0}, v_1, \dots, v_k \in A, t_1 \dots, t_k \in \mathbb{R}_{\ge 0}, \sum_{i=1}^{k} t_i = 1 \right\}$$
, (A.1.2.1)

AConv
$$A = \left\{ \sum_{i=1}^{k} r_i v_i \in E \mid k \in \mathbb{N}_{>0}, v_1, \dots, v_k \in A, r_1 \dots, r_k \in R, \sum_{i=1}^{k} |r_i| \leq 1 \right\}$$
 (A.1.2.2)

For the closed hulls one has

 $\overline{\operatorname{Conv}} A = \overline{\operatorname{Conv} A}$ and $\overline{\operatorname{AConv}} A = \overline{\operatorname{AConv} A}$.

Finally, if A is circled, then

$$AConv A = Conv A .$$

Proof. By definition, the right hand side of Eq. (A.1.2.1) is convex and contains A, hence it contains Conv A. Conversely, one shows by induction on $k \in \mathbb{N}_{>0}$ and convexity of Conv A that each element of the form $\sum_{i=1}^{k} t_i v_i$ with $v_1, \ldots, v_k \in A$ and $t_1, \ldots, t_k \in \mathbb{R}_{\geq 0}$ such that $\sum_{i=1}^{k} t_i = 1$ is in Conv A. This proves Eq. (A.1.2.1). The proof of Eq. (A.1.2.2) is similar. Observe that the right hand side of Eq. (A.1.2.2) is absolutely convex and contains A. Hence it contains AConv A. An argument

using induction on $k \in \mathbb{N}_{>0}$ and absolute convexity of AConv A shows that each element of the form $\sum_{i=1}^{k} r_i v_i$ with $v_1, \ldots v_k \in A$ and $r_1 \ldots, r_k \in R$ such that $\sum_{i=1}^{k} |r_i| \leq 1$ is in Conv A. So Eq. (A.1.2.2) holds true as well. The claim about the closed hulls is a consequence of Lemma 1.2.23. For the proof of the last claim it suffices to show that Conv A is circled if A is. To this end let $v \in \text{Conv } A$ and $r \in R$ with $|r| \leq 1$. Then one can write v in the form $v = \sum_{i=1}^{k} t_i v_i$ with $v_1, \ldots, v_k \in A$ and $t_1, \ldots, t_k \in \mathbb{R}_{\geq 0}$, where $\sum_{i=1}^{k} t_i = 1$. Hence $rv = \sum_{i=1}^{k} t_i (rv_i)$, which is in Conv A, since $rv_i \in A$ for all i because A is circled.

1.2.27 Lemma Let $A \subset E$ be a non-empty subset of a tvs E over a normed real division algebra R. (i) If A is convex and $t_1, \ldots, t_k \in \mathbb{R}_{\geq 0}$ with $k \in \mathbb{N}_{>0}$, then

$$\sum_{i=1}^{k} t_i A = \left(\sum_{i=1}^{k} t_i\right) A \; .$$

(ii) If A is absolutely convex and $r_1, \ldots, r_k \in R$ with $k \in \mathbb{N}_{>0}$, then

$$\sum_{i=1}^k r_i A = \left(\sum_{i=1}^k |r_i|\right) A \; .$$

Proof. ad (i). Obviously $\sum_{i=1}^{k} t_i A \supset \left(\sum_{i=1}^{k} t_i\right) A$. Let us show the converse inclusion. Without loss of generality we can assume that $t_i > 0$ for all i. Then $t = \sum_{i=1}^{k} t_i > 0$, so, after division by t, we can reduce the claim to showing that $\sum_{i=1}^{k} t_i A \subset A$ for $t_1, \ldots, t_k \in \mathbb{R}_{>0}$ such that $\sum_{i=1}^{k} t_i = 1$. But $\sum_{i=1}^{k} t_i A \subset Conv A = A$ by Lemma 1.2.26 and convexity of A.

ad (ii). Since by absolute convexity $r_i A = |r_i| A$ for i = 1, ..., k, the claim follows from (i).

1.2.28 Lemma Let \mathbb{K} be one of the division rings \mathbb{C} or \mathbb{H} with their standard absolute values and let \mathbb{E} be a vector space over \mathbb{K} . Then a convex subset $C \subset \mathbb{E}$ is absorbent in \mathbb{E} if and only if it is absorbent in the realification $\mathbb{E}^{\mathbb{R}}$.

Proof. It suffices to show the non-trivial direction. So assume that C is convex and absorbent in the realification $\mathbb{E}^{\mathbb{R}}$. Denote by u_1, \ldots, u_n the standard basis of \mathbb{K} over \mathbb{R} with n = 2 or n = 4 depending on \mathbb{K} . In particular this means $u_1 = 1$. For given $v \in \mathbb{E}$ there now exists $t \in \mathbb{R}_{\geq 0}$ such that

$$\pm \frac{1}{u_1}v, \dots, \pm \frac{1}{u_n}v \in rC$$
 for all $r \ge t$.

Without loss of generality we can assume $t \ge 1$. Let $z \in \mathbb{K}$ with $|z| \ge nt$. Then the vectors $c_1 = \operatorname{sgn} z_1 \frac{n}{|z| u_1} v, \ldots, c_n = \operatorname{sgn} z_n \frac{n}{|z| u_n} v$ are elements of C. By convexity of C and since $0 \in C$ one has $\frac{|z_1|}{|z|} c_1, \ldots, \frac{|z_n|}{|z|} c_n \in C$. Again by convexity one concludes

$$\frac{1}{z}v = \sum_{i=1}^{n} \frac{z_i}{|z|^2 u_i} v = \sum_{i=1}^{n} \frac{|z_i|}{n|z|} c_i \in C .$$

Hence C is absorbing and the claim is proved.

1.2.29 Definition A topological vector space E over a normed real division algebra R for which Axiom LCVS below holds true is called a *locally convex topological vector space*, a *locally convex vector space* or shortly a *locally convex tvs*.

(LCVS) The vector space topology on E has a base consisting of convex sets.

1.2.30 Remark For better readability, we often say *locally convex topology* instead of *locally convex vector space topology*.

1.2.31 Proposition The locally convex topological vector spaces over a normed real division algebra R together with the continuous linear maps between them form a full subcategory of the category R-TVS of topological R-vector spaces. It is denoted R-LCVS.

Proof. This is clear by definition.

1.2.32 Proposition and Definition The filter of zero neighborhoods of a locally convex topological vector space E over a normed real divison algebra R has a filter base \mathcal{B} with the following properties:

- (i) For each $V \in \mathcal{B}$ there exists $U \in \mathcal{B}$ such that $U + U \subset V$.
- (ii) Every element of \mathcal{B} is a barrel that means is absolutely convex, closed and absorbing.

(iii) Let $r \in R^{\times}$. Then $V \in \mathcal{B}$ if and only if $rV \in \mathcal{B}$.

Conversely, if \mathcal{B} is a filter base on an *R*-vector space *E* such that (i) holds true and such that each element of \mathcal{B} is absolutely convex and absorbing, then there exists a unique locally convex topology on *E* such that \mathcal{B} is a neighborhood base of the origin. It is the coarsest among all translation invariant topologies for which \mathcal{B} is a set of zero neighborhoods and is called the locally convex topology generated or induced by \mathcal{B} .

Proof. Let E be a locally convex tvs. Let \mathcal{B} be the collection of all barrels which are at the same time zero neighborhoods. Let V be an element of \mathcal{U}_0 , the filter of zero neighborhoods. Since E is (T3) by Proposition 1.2.3, there exists a closed zero neighborhood V_a such that $V_a \subset V$. By local convexity of E there exists a convex zero neighborhood V_b with $V_b \subset V_a$. By Proposition 1.2.13 there exists a circled zero neighborhood V_c with $V_c \subset V_b$. The closed convex hull $U = \overline{\text{Conv}} V_c$ then is a barrel contained in V. Since it is a zero neighborhood it is an element of \mathcal{B} , and \mathcal{B} is a filter base of \mathcal{U}_0 . This proves (ii).

To verify (i), let $V \in \mathcal{B}$ and observe that by continuity of addition there exist zero neighborhoods U_1 and U_2 such that $U_1 + U_2 \subset V$. Choose $U \in \mathcal{B}$ such that $U \subset U_1 \cap U_2$. Then $U + U \subset V$.

Claim (iii) holds true since multiplication by an element $r \in R^{\times}$ is a homeomorphism which preserves circled and convex sets.

The remaining claim follows immediately from Proposition 1.2.13 and the observation that a real division algebra is archimedean. $\hfill\square$

1.2.33 Corollary Let S be a non-empty set of absolutely convex and absorbent subsets of a vector space E over a normed real divison algebra R. Then the set

$$\mathcal{B} = \left\{ r \bigcap_{B \in \mathcal{F}} B \in \mathcal{P}(\mathbf{E}) \mid \mathcal{F} \in \mathcal{P}_{\mathsf{fin}}(\mathcal{S}), \ \mathcal{F} \neq \emptyset \ \& \ r \in R^{\times} \right\}$$

consists of absolutely convex and absorbent subsets of V and is a base of the filter of zero neighborhoods of a locally convex topology T on E uniquely determined by that property. This topology is the coarsest among all vector space topologies for which S is a set of zero neighborhoods. The topology T is called the locally convex topology generated or induced by S.

Proof. The intersection of finitely many absolutely convex and absorbing sets is non-empty and again absolutely convex and absorbing by Lemma 1.2.12 (i) and Proposition and Definition 1.2.24. By Lemma 1.2.12 (ii) and Lemma 1.2.23, the scalar multiple of an absolutely convex and absorbing set again has these properties whenever the scalar is invertible. Hence each element of \mathcal{B} is absolutely convex and absorbing. Given two elements $C, D \in \mathcal{B}$ there exist non-empty $\mathcal{F}, \mathcal{G} \in \mathcal{P}_{fin}(\mathcal{S})$ and $r, s \in \mathbb{R}^{\times}$ such that $C = r \bigcap_{B \in \mathcal{F}} B$ and $D = s \bigcap_{B \in \mathcal{G}} B$. Without loss of generality one can assume that $|r| \leq |s|$. Then $A = r \bigcap_{B \in \mathcal{F} \cup \mathcal{G}} B \in \mathcal{B}$ and $A = C \cap rs^{-1}D \subset C \cap D$ since D is balanced and $|rs^{-1}| \leq 1$. Hence \mathcal{B} is a filter base consisting of absolutely convex and absorbent sets. Moreover, $\frac{1}{2}C + \frac{1}{2}C \subset C$ for every $C \in \mathcal{B}$ by absolut convexity. By Proposition 1.2.32 the filter base \mathcal{B} therefore generates a unique locally convex topology \mathcal{T} for which \mathcal{B} is a base of the filter of zero neighborhoods. Moreover, \mathcal{T} is the coarsest translation invariant topology so that \mathcal{B} is a set of zero neighborhoods. This implies in particular that \mathcal{S} is a zero neighborhood. Then finite intersections of elements of \mathcal{S} are zero neighborhoods with respect to \mathcal{T}' and therefore also all elements of \mathcal{B} . Since \mathcal{T}' is translation invariant one concludes that \mathcal{T} is coarser than \mathcal{T}' and the claim is proved.

A.1.3. Seminorms and gauge functionals

1.3.1 Throughout the rest of this chapter the symbol \mathbb{K} will always stand for the field of real numbers \mathbb{R} , the field of complex numbers \mathbb{C} or the division algebra of quaternions \mathbb{H} . We assume these division algebras to be equipped with their standard absolute values $|\cdot|$. Moreover, vector spaces are assumed to be defined over the ground field \mathbb{K} unless mentioned differently and are always assumed to be left vector spaces.

Seminorms and induced vector space topologies

1.3.2 Definition By a *seminorm* on a vector space E one understands a map $p : E \to \mathbb{R}$ with the following properties:

(N0) The map p is positive that is $p(v) \ge 0$ for all $v \in E$.

(N1) The map p is absolutely homogeneous that means

$$p(rv) = |r| p(v)$$
 for all $v \in E$ and $r \in \mathbb{K}$.

(N2) The map p is subadditive or in other words satisfies the triangle inequality

 $p(v+w) \leq p(v) + p(w)$ for all $v, w \in E$.

A seminorm is called a *norm* if in addition the following axiom is satisfied:

(N3) For all $v \in E$ the relation p(v) = 0 holds true if and only if v = 0.

A vector space E equipped with a norm $\|\cdot\|: E \to \mathbb{R}_{\geq 0}$ is called a *normed vector space*.

1.3.3 Let us introduce some useful further properties a map $p : E \to \mathbb{R}$ can have. One calls such a map p

- (1) positively homogeneous if p(tv) = t p(v) for all $t \in \mathbb{R}_{>0}$ and all $v \in \mathbb{E}$,
- (2) sublinear if $p(tv + sw) \leq t p(v) + s p(w)$ for all $t, s \in \mathbb{R}_{\geq 0}$ and all $v, w \in E$, and
- (3) convex if $p(tv + sw) \leq t p(v) + s p(w)$ for all $t, s \in \mathbb{R}_{\geq 0}$ with t + s = 1 and all $v, w \in E$.

1.3.4 Lemma For a real-valued map $p : E \to \mathbb{R}$ on a vector space E the following are equivalent:

- (i) p is sublinear.
- (ii) *p* is positively homogeneous and convex.
- (iii) *p* is positively homogeneous and subadditive.

Proof. Let p be sublinear. Then p is subadditive by definition. Subadditivity implies $p(0) \le p(0) + p(0)$, hence $p(0) \ge 0$. By sublinearity

$$p(0) = p(0 \cdot 0 + 0 \cdot 0) \le 0 \cdot p(0) + 0 \cdot p(0) = 0$$

so p(0) = 0. We show that p is positively homogeneous. Applying sublinearity again one checks for $v \in E$ and $t \ge 0$ that

$$p(tv) = p(tv + 0 \cdot 0) \le tp(v) + 0 \cdot p(0) = tp(v) ,$$

so p is positively homogeneous and the implication (i) \implies (iii) follows. If p is positively homogeneous and subadditive, then for $v, w \in E$ and t, s > 0 with t + s = 1

$$p(tv + sw) \leq p(tv) + p(sw) \leq tp(v) + sp(w),$$

so p is convex. This gives the implication (iii) \implies (ii). If p is positively homogeneous and convex, then one computes for $v, w \in E$ and $t, s \ge 0$ with t + s > 0

$$p(tv+sw) = (t+s)p\left(\frac{t}{t+s}v + \frac{s}{t+s}w\right) \leq (t+s)\left(\frac{t}{t+s}p(v) + \frac{s}{t+s}p(w)\right) = tp(v) + sp(w) .$$

Since $p(0) = \lim_{t \searrow 0} p(t0) = \lim_{t \searrow 0} t p(0) = 0$ by positive homogeneity, p then has to be sublinear and one obtains the implication (ii) \implies (i).

1.3.5 Lemma Let $p : E \to \mathbb{R}$ be a real-valued map defined on a vector space E over K.

- (i) If $p : E \to \mathbb{R}$ is positively homogeneous, then p(0) = 0.
- (ii) If $p : E \to \mathbb{R}$ is subadditive, then $p(0) \ge 0$ and for all $v, w \in E$

$$|p(v) - p(w)| \le \max\{p(v - w), p(w - v)\}$$
.

- (iii) If $p : E \to \mathbb{R}$ is convex, then the sets $\mathbb{B}_{p,\varepsilon} := \{v \in E \mid p(v) < \varepsilon\}$ and $\overline{\mathbb{B}}_{p,\varepsilon} := \{v \in E \mid p(v) \leq \varepsilon\}$ are convex for all $\varepsilon > 0$.
- (iv) If p is sublinear, then $\mathbb{B}_{p,\varepsilon}$ and $\overline{\mathbb{B}}_{p,\varepsilon}$ are convex and absorbent for all $\varepsilon > 0$.

Proof. ad (i). As already observed, $p(0) = \lim_{t \searrow 0} p(t0) = \lim_{t \searrow 0} t p(0) = 0$.

ad (ii). Note that by subadditivity

$$p(0)\leqslant p(0)+p(0),\quad p(v)-p(w)\leqslant p(v-w),\quad \text{and}\quad p(w)-p(v)\leqslant p(w-v)\ .$$

This entails (ii).

ad (iii). Let $v, w \in \{v \in E \mid p(v) < \varepsilon\}$ and $0 \leq t \leq 1$. Then, by convexity of p,

$$p(tv + (1-t)w) \leq tp(v) + (1-t)p(w) < t\varepsilon + (1-t)\varepsilon = \varepsilon$$

Hence $tv + (1 - t)w \in \{v \in E \mid p(v) < \varepsilon\}$. The proof for $\{v \in E \mid p(v) \leq \varepsilon\}$ is analogous.

ad (iv). Convexity of the sets $\mathbb{B}_{p,\varepsilon}$ and $\overline{\mathbb{B}}_{p,\varepsilon}$ holds by (iii). Moreover, $\mathbb{B}_{p,\varepsilon} \subset \overline{\mathbb{B}}_{p,\varepsilon}$ by definition. Hence it suffices by Lemma 1.2.28 to show that $\mathbb{B}_{p,\varepsilon}$ is absorbent in the realification $\mathbb{E}^{\mathbb{R}}$. Since p is positively homogenous by Lemma 1.3.4 and $0 \leq p(v) + p(-v)$ for all $v \in \mathbb{E}$, one concludes that for all $t \in \mathbb{R}$ and $v \in \mathbb{E}$

$$|p(tv)| \leq |t| \max\{p(v), p(-v)\} .$$

Hence $tv \in \mathbb{B}_{p,\varepsilon}$ if $0 < t < \frac{\varepsilon}{\max\{p(v), p(-v)\}+1}$, and $\mathbb{B}_{p,\varepsilon}$ is absorbent in $\mathbb{E}^{\mathbb{R}}$. \Box

1.3.6 Definition If $p : E \to \mathbb{R}$ is a seminorm on a vector space E, we denote for every $v \in E$ and $\varepsilon > 0$ by $\mathbb{B}_{p,\varepsilon}(v)$ the (open) ε -ball associated with p and with center v that is the set

$$\mathbb{B}_{p,\varepsilon}(v) = \left\{ w \in \mathbf{E} \mid p(w-v) < \varepsilon \right\} \,.$$

The closed ε -ball associated with p and with center v is defined as

$$\overline{\mathbb{B}}_{p,\varepsilon}(v) = \left\{ w \in \mathbf{E} \mid p(w-v) \leqslant \varepsilon \right\} \,.$$

The positive number ε is called the *radius* of the ball. In case the center of the ball is the origin, we often write $\mathbb{B}_{p,\varepsilon}$ and $\overline{\mathbb{B}}_{p,\varepsilon}$ for $\mathbb{B}_{p,\varepsilon}(0)$ and $\overline{\mathbb{B}}_{p,\varepsilon}(0)$, respectively. If in addition the radius equals 1, then we usually write only \mathbb{B}_p and $\overline{\mathbb{B}}_p$ and call these sets the *open* respectively the *closed unit ball*. More generally, for the particular radius 1 we denote the corresponding balls by $\mathbb{B}_p(v)$ and $\overline{\mathbb{B}}_p(v)$ and call them the *open* respectively *closed unit balls with center* v. When by the context it is clear which seminorm p a ball is associated with we often do not mention p explicitly. This is in particular the case when the underlying vector space is a normed vector space.

If P is a finite set or a finite family of seminorms on E we define the open and closed ε -multiballs with center v by

$$\mathbb{B}_{P,\varepsilon}(v) = \left\{ w \in \mathcal{E} \mid p(w-v) < \varepsilon \text{ for all } p \in P \right\}$$

and

$$\overline{\mathbb{B}}_{P,\varepsilon}(v) = \left\{ w \in \mathbf{E} \mid p(w-v) \leqslant \varepsilon \text{ for all } p \in P \right\} \,,$$

respectively. As before, we abbreviate $\mathbb{B}_{P,\varepsilon} = \mathbb{B}_{P,\varepsilon}(0)$ and $\overline{\mathbb{B}}_{P,\varepsilon} = \overline{\mathbb{B}}_{P,\varepsilon}(0)$.

1.3.7 Remark For convenience, we will also use the symbols $\mathbb{B}_{p,\varepsilon}$ and $\overline{\mathbb{B}}_{p,\varepsilon}$ to denote the sets $\{v \in E \mid p(v) < \varepsilon\}$ and $\{v \in E \mid p(v) \leqslant \varepsilon\}$, respectively, when $p : E \to \mathbb{R}$ is just a real-valued convex map on the vector space E. Note that for such a p the set $\{v \in E \mid p(v) < 0\}$ might be non-empty. But as we have shown in Lemma 1.3.5 the sets $\mathbb{B}_{p,\varepsilon}$ and $\overline{\mathbb{B}}_{p,\varepsilon}$ associated to a convex p share with the the balls associated to a seminorm several nice properties like convexity.

1.3.8 Proposition Let E be a \mathbb{K} -vector space, and P a finite set of seminorms on E. Then, for every $\varepsilon > 0$ and $v \in E$, the ε -multiballs $\mathbb{B}_{P,\varepsilon}(v)$ and $\overline{\mathbb{B}}_{P,\varepsilon}(v)$ are convex. The ε -multiballs $\mathbb{B}_{P,\varepsilon}$ and $\overline{\mathbb{B}}_{P,\varepsilon}$ centered at the origin are absolutely convex and absorbent.

Proof. Axiom (N1) immediately entails that $\mathbb{B}_{P,\varepsilon}$ and $\overline{\mathbb{B}}_{P,\varepsilon}$ are circled. Axiom (N2) together with (N1) entails that the sets $\mathbb{B}_{P,\varepsilon}(v)$ and $\overline{\mathbb{B}}_{P,\varepsilon}(v)$ are convex. Namely, if $w_1, w_2 \in \mathbb{B}_{P,\varepsilon}(v)$ and $t \in [0,1]$, then one has for all seminorms $p \in P$

$$p(tw_1 + (1-t)w_2 - v) \le t p(w_1 - v) + (1-t) p(w_2 - v) < t \varepsilon + (1-t) \varepsilon = \varepsilon$$

and likewise $p(tw_1 + (1-t)w_2 - v) \leq \varepsilon$ for all $w_1, w_2 \in \overline{\mathbb{B}}_{P,\varepsilon}(v)$ and $p \in P$.

Now let $v \in E$ and $\varepsilon > 0$ be given. Put $t_p = \frac{p(v)+1}{\varepsilon}$ for every $p \in P$ and $t_0 = \max\{t_p \mid p \in P\}$. Then one has for all $t \in \mathbb{K}$ with $|t| \ge t_0$ and for all $p \in P$

$$p\left(\frac{1}{t}v\right) \leq \frac{\varepsilon}{p(v)+1} p(v) < \varepsilon$$
,

hence $v \in t \mathbb{B}_{P,\varepsilon}$. So $\mathbb{B}_{P,\varepsilon}$ is absorbing. Since $\mathbb{B}_{P,\varepsilon}$ contains the absorbing set $\mathbb{B}_{P,\varepsilon}$, it is absorbing as well.

1.3.9 Proposition and Definition Assume to be given a set Q of seminorms on a vector space E. Let $\mathcal{P}_{fin}(Q)$ be the collection of all finite subsets of Q. A base of a topology on E then is given by

$$\mathcal{B} = \{ \mathbb{B}_{P,\varepsilon}(v) \mid P \in \mathcal{P}_{\mathsf{fin}}(Q), v \in \mathcal{E}, \varepsilon > 0 \} .$$

The topology \mathcal{T} generated by \mathcal{B} is called the topology generated, induced or defined by Q. Moreover, \mathcal{T} is a locally convex vector space topology on \mathcal{E} . It coincides with the coarsest translation invariant topology on \mathcal{E} such that each seminorm in Q is continuous.

Proof. Consider the set \mathcal{B}_0 of all multiballs $\mathbb{B}_{P,\varepsilon}$ with $P \in \mathcal{P}_{fin}(Q)$ and $\varepsilon > 0$ centered at the origin. Clearly, \mathcal{B}_0 is a filter base since for $P_1, P_2 \in \mathcal{P}_{fin}(Q)$ and $\varepsilon_1, \varepsilon_2 > 0$ the multiball $\mathbb{B}_{P_1 \cup P_2, \min\{\varepsilon_1, \varepsilon_2\}}$ is contained in $\mathbb{B}_{P_1,\varepsilon_1} \cap \mathbb{B}_{P_2,\varepsilon_2}$. Moreover it consists of absolutely convex and absorbing sets by Proposition 1.3.8.

By a similar argument one shows that \mathcal{B} is base of a topology. Let $\mathbb{B}_{P_1,\varepsilon_1}(v_1)$, $\mathbb{B}_{P_2,\varepsilon_2}(v_2) \in \mathcal{B}$ and $v \in \mathbb{B}_{P_1,\varepsilon_1}(v_1) \cap \mathbb{B}_{P_2,\varepsilon_2}(v_2)$. Let ε be the minium of the numbers $\varepsilon_1 - p_1(v - v_1)$ and $\varepsilon_2 - p_2(v - v_2)$, where p_1 runs through the elements of P_1 and p_2 through the ones of P_2 . Then $\varepsilon > 0$ and $\mathbb{B}_{P_1 \cup P_2,\varepsilon}(v) \subset \mathbb{B}_{P_1,\varepsilon_1}(v_1) \cap \mathbb{B}_{P_2,\varepsilon_2}(v_2)$, and \mathcal{B} is a base for a topology \mathcal{T} indeed. By construction, \mathcal{B}_0 then is a base for the filter of zero neighborhoods and each element of \mathcal{B}_0 is open in \mathcal{T} . Moreover, each closed multiball $\overline{\mathbb{B}}_{P,\varepsilon}(v)$ is closed in \mathcal{T} since the complement $E \setminus \overline{\mathbb{B}}_{P,\varepsilon}(v)$ contains with w also the open multiball $\mathbb{B}_{P,\delta}(w)$, where $\delta = \min\{p(v - w) - \varepsilon | p \in P\}$.

We now prove continuity of addition with respect to T. Let $v_1, v_2 \in E$, $P \in \mathcal{P}_{fin}(Q)$, and $\varepsilon > 0$. Since the triangle inequality holds for every seminorm in F, one has

$$\mathbb{B}_{P,\frac{\varepsilon}{2}}(v_1) + \mathbb{B}_{P,\frac{\varepsilon}{2}}(v_2) \subset \mathbb{B}_{P,\varepsilon}(v_1 + v_2) + \mathbb{B}_{P,\varepsilon}(v_1 + v_2) = \mathbb{B}_{P,\varepsilon}(v_1 + v_2) + \mathbb{B}$$

which entails continuity of addition at each $(v_1, v_2) \in E \times E$. Next consider multiplication by scalars and let $\lambda \in \mathbb{K}$ and $v \in E$. Again let $P = \{p_1, \ldots, p_n\} \in \mathcal{P}_{fin}(Q)$ and $\varepsilon > 0$. Let $C_1 = \sup\{p_j(v) \mid 1 \leq j \leq n\} + 1$, $C_2 = |\lambda| + 1$ and put $\delta_1 = \min\{1, \frac{\varepsilon}{2C_1}\}$ and $\delta_2 = \frac{\varepsilon}{2C_2}$. Then one obtains by absolute homogeneity and subadditivity of each seminorm

$$p_j(\mu w - \lambda v) \leq |\mu| p_j(w - v) + |\mu - \lambda| p_j(v)$$
 for all $\mu \in \mathbb{K}$ and $w \in \mathbb{E}$,

hence

$$\mathbb{B}_{\delta_1}(\lambda) \cdot \mathbb{B}_{P,\delta_2}(v) \subset \mathbb{B}_{P,\varepsilon}(\lambda \cdot v) ,$$

where $\mathbb{B}_{\delta_1}(\lambda) = \{\mu \in \mathbb{K} \mid |\mu - \lambda| < \delta_1\}$. This shows continuity of scalar multiplication at each $(\lambda, v) \in \mathbb{K} \times E$, and \mathcal{T} is a vector space topology.

Since each of the base elements $\mathbb{B}_{P,\varepsilon} \in \mathcal{B}_0$ is convex, Axiom LCVS holds true as well and the topology \mathcal{T} is locally convex.

Every seminorm $p \in Q$ is continuous with respect to the topology \mathcal{T} since for all a < b the preimage $p^{-1}((a, b)) = \mathbb{B}_{p,b} \setminus \overline{\mathbb{B}}_{p,a}$ is open in \mathcal{T} . Now let \mathcal{T}' be a translation invariant topology on \mathcal{E} for which every seminorm $p \in Q$ is continuous. In that topology \mathcal{B}_0 is a set of zero neighborhoods. As shown before, every element $B \in \mathcal{B}_0$ is absolutely convex, absorbing and satisfies $\frac{1}{2}B + \frac{1}{2}B \subset B$. Hence by Proposition and Definition 1.2.32 the topology \mathcal{T}' is finer than the locally convex topology generated by \mathcal{B}_0 . But the latter topology coincides with \mathcal{T} by construction. This shows the last part of the claim and the proof is finished.

Gauge functionals and induced seminorms

1.3.10 As we have seen, any vector space with a topology defined by a family of seminorms on it is a locally convex topological vector space. The converse also holds true. The fundamental notion needed for the proof of this is the following.

1.3.11 Definition Let E be a vector space and $A \subset E$ absorbent. Then the map

$$p_A : \mathbf{E} \to \mathbb{R}_{\geq 0}, v \mapsto p_A(v) = \inf \{ t \in \mathbb{R}_{>0} \mid v \in tA \}$$

is called the gauge functional, the Minkowski functional or the Minkowski gauge of A.

1.3.12 Remark By definition of an absorbent set, $\{t \in \mathbb{R}_{>0} \mid v \in tA\}$ is non-empty whenever $A \subset E$ is absorbent. Hence p_A is well-defined for such A.

1.3.13 Proposition The Minkowski gauge $p_A : E \to \mathbb{R}_{\geq 0}$ of an absorbent subset A of a vector space E has the following properties.

(i) The gauge functional is positively homogeneous that is $p_A(tv) = t p_A(v)$ for all $t \in \mathbb{R}_{>0}$ and all $v \in \mathbb{E}$.

(ii) If A is convex, then p_A is subadditive and

$$\mathbb{B}_p(v) = \bigcup_{0 < t < 1} tA \subset A \subset \bigcap_{1 < t} tA = \overline{\mathbb{B}}_p(v)$$

(iii) If A is absolutely convex, then p_A is a seminorm on E.

Proof. If t > 0, then $tv \in sA$ for some s > 0 if and only if $v \in \frac{s}{t}A$. Hence $\{s \in \mathbb{R}_{>0} \mid tv \in sA\}$ and $t\{s \in \mathbb{R}_{>0} \mid v \in sA\}$ coincide for all t > 0, so (i) follows.

Assume that A is convex. Let $v, w \in E$ and $\varepsilon > 0$. Then there exist $t > p_A(v)$ and $s > p_A(w)$ such that $v \in tA$, $w \in sA$, $t < p_A(v) + \frac{\varepsilon}{2}$ and $s < p_A(w) + \frac{\varepsilon}{2}$. By convexity of A and Lemma 1.2.27, $v + w \in tA + sA = (t + s)A$. Hence $p_A(v + w) \leq (t + s) < p_A(v) + p_A(w) + \varepsilon$. Since $\varepsilon > 0$ was arbitrary, $p_A(v + w) \leq p_A(v) + p_A(w)$ and p_A is subadditive. If $v \in tA$ for some t with 0 < t < 1, then $p_A(v) \leq t < 1$ by definition. Conversely, if $p_A(v) < 1$, then there exists a t > 0 such that t < 1 and $v \in tA$. Hence the equality $\mathbb{B}_p(v) = \bigcup_{0 < t < 1} tA$ follows. Since A is absorbing, 0 is an element of A. By convexity of A one therefore concludes $tA = (1 - t)\{0\} + tA \subset A$ whenever 0 < t < 1. For t > 1 this shows $\frac{1}{t}A \subset A$, hence $A \subset tA$. So the relation $\bigcup_{0 < t < 1} tA \subset A \subset \bigcap_{1 < t} tA$ is proved. Now assume that $v \in tA$ for all t > 1. Then $p_A(v) \leq 1$ by definition. If conversely $p_A(v) \leq 1$, then there exists for each $\varepsilon > 0$ an $s \geq 0$ such that $p_A(v) \leq s$, $v \in sA$ and $s < 1 + \varepsilon$. Hence, for $t \geq 1 + \varepsilon$ by Lemma 1.2.27 and $0 \in A$,

$$v \in sA = sA + (t-s)\{0\} \subset sA + (t-s)A = tA$$
.

Since $\varepsilon > 0$ was arbitrary, $v \in tA$ for all t > 1 follows. So one obtains the equality $\bigcap_{1 < t} tA = \mathbb{B}_p(v)$, and (ii) is proved.

To verify (iii) recall that A is circled whenever A is absolutely convex. This entails for $r \in \mathbb{K}$, $v \in E$ and absolutely convex A

$$p_A(rv) = \inf \left\{ t \in \mathbb{R}_{>0} \mid rv \in tA \right\} = \inf \left\{ t \in \mathbb{R}_{>0} \mid |r|v \in tA \right\} = p_A(|r|v) = |r|p_A(v) ,$$

where for the last equality we have used (i).

1.3.14 Lemma Let A and B be absorbent subsets of a vector space E. Then the following holds true.

- (i) $p_{tA}(v) = p_A(t^{-1}v)$ for all $t \in \mathbb{K}^{\times}$ and $v \in \mathbb{E}$.
- (ii) If $B \subset A$, then $p_A \leq p_B$.
- (iii) If A is convex, then $v \in tA$ for all $v \in E$ and $t > p_A(v)$.
- (iv) If A and B are convex, then the intersection $A \cap B$ is absorbent and convex and $p_{A \cap B} = \sup\{p_A, p_B\}$, where $\sup\{p_A, p_B\}(v) = \sup\{p_A(v), p_B(v)\}$ for all $v \in E$.

Proof. ad (i). If $t \in \mathbb{K}$ is invertible, then $v \in tA$ if and only if $t^{-1}v \in A$.

ad (ii). Let $v \in E$ and $\varepsilon > 0$. Then there exists t with $p_B(v) \le t < p_B(v) + \varepsilon$ such that $v \in tB$. By $B \subset A$ this implies $v \in tA$, hence $p_A(v) \le t < p_B(v) + \varepsilon$. Since $\varepsilon > 0$ was arbitrary, the estimate $p_A \le p_B$ follows.

ad (iii). By definition of the Minkowski gauge there exists $s \in \mathbb{R}$ such that $p_A(v) < s < t$ and $v \in sA$. By convexity of A one concludes $\frac{s}{t}v = \frac{s}{t}v + (1 - \frac{s}{t}) \cdot 0 \in sA$, hence $v \in tA$.

ad (iv). The intersection of convex sets is convex, so $A \cap B$ is convex. Let $v \in E$ and choose $r_A \ge 0$ and $r_B \ge 0$ such that $v \in tA$ for all $t \ge r_A$ and $v \in sB$ for all $s \ge r_B$. Then $v \in (tA) \cap (tB) = t(A \cap B)$ for all $t \ge \max\{r_A, r_B\}$, so $A \cap B$ is absorbent. One has $p_{A \cap B} \ge \sup\{p_A, p_B\}$ by (ii). To show the converse inequality assume that $v \in E$ and $t > \sup\{p_A(v), p_B(v)\}$. Then $v \in tA \cap tB = t(A \cap B)$, which implies $p_{A \cap B}(v) \le t$. Hence $p_{A \cap B}(v) \le \sup\{p_A(v), p_B(v)\}$ since $t > \sup\{p_A(v), p_B(v)\}$ was arbitrary.

1.3.15 Lemma Let $p : E \to \mathbb{R}$ be a sublinear map on a vector space E and $A \subset E$ convex. If

$$\mathbb{B}_p \subset A \subset \overline{\mathbb{B}}_p \; ,$$

then the gauge functional p_A coincides with $\sup\{p, 0\}$. If p is even a seminorm, then $p = p_A$.

Proof. Let $p : E \to \mathbb{R}$ be sublinear. Observe that then \mathbb{B}_p is absorbent by Lemma 1.3.5 (iv). Hence A must also be absorbent by assumption, so the associated Minkowski gauge p_A is positively homogeneous by Proposition 1.3.13 (i).

Assume now that there exists $v \in E$ such that $\max\{p(v), 0\} < p_A(v)$. By positive homogeneity of p and p_A one can achive by possibly multiplying v by a positive real number that $\max\{p(v), 0\} < 1 < p_A(x)$. The first inequality entails $v \in \mathbb{B}_p$, the second $v \notin \overline{\mathbb{B}}_p$ which is a contradiction. Next assume that there exists $v \in E$ with $p_A(v) < \max\{p(v), 0\}$. As before one can then achieve that $p_A(v) < 1 < \max\{p(v), 0\}$ for some $v \in E$. By the first inequality one concludes $v \in A$, by the second $v \notin A$. This is a contradiction. So the equality $\max\{p(v), 0\} = p_A(v)$ holds for all $v \in E$.

In case p is a seminorm, then $p(v) \ge 0$ for all $v \in E$ and the second claim follows by the first. \Box

1.3.16 Proposition Let E be a topological vector space, and $p : E \to \mathbb{R}$ be sublinear. Then the following are equivalent.

- (i) The map p is continuous in the origin.
- (ii) The map p is uniformly continuous.
- (iii) The map p is continuous.
- (iv) The unit ball \mathbb{B}_p is a zero neighborhood.

Proof. Let us first show (i) \implies (ii). To this end fix $\varepsilon > 0$. By assumption there exists a zero neighborhood $V \subset E$ such that $|p(v)| < \varepsilon$ for all $v \in V$. By possibly passing to $V \cap (-V)$ one can assume that V is symmetric. Lemma 1.3.5 (ii) now implies

$$|p(v) - p(w)| < \varepsilon$$
 for all $v, w \in V$.

Hence p is uniformly continuous. The implications (ii) \implies (iii) and (iii) \implies (iv) are trivial. It remains to prove (iv) \implies (i). Assume that $\mathbb{B}_p(0,1)$ is a zero neighborhood. Then there exists a symmetric zero neighborhood V contained in $\mathbb{B}_p(0,1)$. Since p(0) = 0 one concludes by Lemma 1.3.5 (ii)

$$p(v)| < \max\{p(v), p(-v)\} < 1$$
 for all $v \in V$.

But this implies $|p(v)| < \varepsilon$ for all $v \in \varepsilon V$ and $\varepsilon > 0$, so p is continuous at the origin.

Normability

1.3.17 Definition A topological vector space E is called *seminormable* if its topology is generated by a single seminorm $p : E \to \mathbb{R}_{\geq 0}$. If the topology on E coincides with the vector space topology generated by a norm $\|\cdot\|$, then one calls E *normable*.

1.3.18 Theorem (Kolmogorov's normability criterion) A topological vector space E is normable if and only if it is a ?? space and possesses a bounded convex neighborhood of the origin.

A.1.4. Function spaces and their topologies

1.4.1 Proposition Let X be a topological space and (Y, d) a metric space. Then the following holds true.

(i) The space

$$\mathcal{B}(X,Y) = \left\{ f : X \to Y \mid \exists y_0 \in Y \exists C > 0 \,\forall x \in X : \, d(f(x), y_0) \leqslant C \right\}$$

of bounded functions from X to Y is a metric space with metric

$$\varrho: \mathcal{B}(X,Y) \times \mathcal{B}(X,Y) \to \mathbb{R}_{\geq 0}, \ (f,g) \mapsto \sup_{x \in X} d(f(x),g(x)) \ .$$

(ii) If (Y, d) is complete, then $(\mathcal{B}(X, Y), \varrho)$ is so, too.

(iii) The space

$$\mathcal{C}_{\mathsf{b}}(X,Y) = \mathcal{C}(X,Y) \cap \mathcal{B}(X,Y)$$

of continuous bounded functions from X to Y is a closed subspace of $\mathcal{B}(X,Y)$.

Proof. Note first that by the triangle inequality there exists for every $f \in \mathcal{B}(X, Y)$ and $y \in Y$ a real number $C_{f,y} > 0$ such that

$$d(f(x), y) \leqslant C_{f,x}$$
 for all $x \in X$.

ad (i). Before verifying the axioms of a metric for ϱ we need to show that ϱ is well-defined meaning that $\sup_{x \in X} d(f(x), g(x)) < \infty$ for all $f, g \in \mathcal{B}(X, Y)$. To this end fix some $y \in Y$ and observe using the triangle inequality that

$$d(f(x),g(x)) \leq d(f(x),y) + d(y,g(x)) \leq C_{f,y} + C_{g,y} \quad \text{for all } x \in X .$$

Since furthermore $d(f(x), g(x)) \ge 0$ for all $x \in X$, the map ϱ is well-defined indeed with image in $\mathbb{R}_{\ge 0}$. If $\varrho(f,g) = 0$, then d(f(x), g(x)) = 0 for all $x \in X$, hence f = g. Obviously, ϱ is symmetric since d is symmetric. Finally, let $f, g, h \in \mathcal{B}(X, Y)$ and check using the triangle inequality for d:

$$\begin{split} \varrho(f,g) &= \sup_{x \in X} d\big(f(x),g(x)\big) \leqslant \sup_{x \in X} \big(d\big(f(x),h(x)\big) + d\big(h(x),g(x)\big)\big) \leqslant \\ &\leqslant \sup_{x \in X} d\big(f(x),h(x)\big) + \sup_{x \in X} d\big(h(x),g(x)\big) = d(f,h) + d(h,g) \;. \end{split}$$

Hence ρ is a metric.

ad (ii). Assume (Y,d) to be complete and let $(f_n)_{n\in\mathbb{N}}$ be a Cauchy sequence in $\mathcal{B}(X,Y)$. Let $\varepsilon > 0$ and choose $N_{\varepsilon} \in \mathbb{N}$ so that

$$\varrho(f_n, f_m) < \varepsilon \quad \text{for all } n, m \ge N$$

Then for every $x \in X$ the relation

$$d(f_n(x), f_m(x)) < \varepsilon \quad \text{for all } n, m \ge N_{\varepsilon} \tag{A.1.4.1}$$

holds true, so $(f_n(x))_{n \in \mathbb{N}}$ is a Cauchy sequence in Y. By completeness of (Y, d) it has a limit which we denote by f(x). By passing to the limit $m \to \infty$ in (A.1.4.1) one obtains that

$$d(f(x), f_n(x)) \leq \varepsilon$$
 for all $x \in X$ and $n \geq N_{\varepsilon}$. (A.1.4.2)

Using the triangle inequality one infers from this for an element $y \in Y$ which we now fix that

$$d(f(x), y)) \leq d(f(x), f_{N_1}(x))) + d(f_{N_1}(x), y) \leq 1 + C_{f_{N_1}, y}$$
.

Hence f is a bounded function. Moreover, (A.1.4.2) entails that

$$\varrho(f, f_n) = \sup_{x \in X} d(f(x), f_n(x)) \leqslant \varepsilon \quad \text{for all } n \ge N_{\varepsilon} ,$$

so $(f_n)_{n \in \mathbb{N}}$ converges to f.

ad (iii). We have to show that the limit f of a sequence $(f_n)_{n\in}$ of functions $f_n \in \mathcal{C}_b(X, Y)$ which converges in $(\mathcal{B}(X, Y), \varrho)$ has to be continuous. To this end let $\varepsilon > 0$ and choose $N_{\varepsilon} \in \mathbb{N}$ so that

$$\varrho(f_n,f) < rac{arepsilon}{3} \quad ext{for all } n \geqslant N_arepsilon \ .$$

Let $x_0 \in X$. By continuity of $f_{N_{\varepsilon}}$ there exists a neighborhood $U \subset X$ of x so that

$$dig(f_{N_arepsilon}(x),f_{N_arepsilon}(x_0)ig) < rac{arepsilon}{3} \quad ext{for all } x \in U \;.$$

By the triangle inequality one concludes that

$$d(f(x), f(x_0)) \leq d(f(x), f_{N_{\varepsilon}}(x)) + d(f_{N_{\varepsilon}}(x), f_{N_{\varepsilon}}(x_0)) + d(f_{N_{\varepsilon}}(x_0), f(x_0)) < \varepsilon$$

for all $x \in U$. Hence f is continuous at x_0 . Since $x_0 \in X$ was arbitrary f, is a continuous map, hence an elemnt of $\mathcal{C}_{\mathbf{b}}(X, Y)$.

1.4.2 Proposition Let X be a topological space and \mathbb{K} the division algebra of real or complex numbers or of quaternions. Then the following holds true.

(i) The space $\mathcal{B}(X,\mathbb{K})$ of bounded \mathbb{K} -valued functions on X can be expressed as

$$\mathcal{B}(X,\mathbb{K}) = \left\{ f: X \to \mathbb{K} \mid \exists C > 0 \ \forall x \in X : \ |f(x)| \leq C \right\}.$$
(A.1.4.3)

It carries the structure of a \mathbb{K} -algebra by pointwise addition and multiplication of functions and becomes a Banach algebra when equipped with the supremums-norm

$$\|\cdot\|_{\infty}: \mathcal{B}(X,\mathbb{K}) \to \mathbb{K}, \quad f \mapsto \sup_{x \in X} |f(x)|.$$

(ii) The subspace C_b(X, K) ⊂ B(X, K) of bounded continuous K-valued functions on X is a closed subalgebra of (B(X, K), ||·||∞), so a Banach algebra as well when endowed with the supremums-norm. For X compact this means in particular that the algebra (C(X, K), ||·||∞) is a Banach algebra.

Proof. Eq. (A.1.4.3) is obvious since the distance of two elements $a, b \in \mathbb{K}$ is given by d(a, b) = |a-b|, so in particular d(a, 0) = |a|. Let $f, g \in \mathcal{B}(X, \mathbb{K})$ and choose $C_f, C_g \ge 0$ so that $|f(x)| \le C_f$ and $|g(x)| \le C_g$ for all $x \in X$. Then, by the triangle inequality and absolute homogeneity of the absolute value,

$$|f(x) + g(x)| \leq C_f + C_g, \quad |a f(x)| \leq |a| C_f, \text{ and } |f(x) \cdot g(x)| \leq C_f \cdot C_g$$

Hence the sum and the product of two bounded functions are bounded and so is any scalar multiple of a bounded function. Therefore, $\mathcal{B}(X,\mathbb{K})$ is an algebra over \mathbb{K} . Using the triangle inequality and absolute homogeneity of the absolute value again one verifies that $||f||_{\infty}$ is a norm on $\mathcal{B}(X,\mathbb{K})$ indeed and that it fulfills $||fg||_{\infty} \leq ||f||_{\infty} \cdot ||g||_{\infty}$ for all $f, g \in \mathcal{B}(X,\mathbb{K})$. Furthermore, by definition, $||f||_{\infty} = \varrho(f,0)$ for all $f \in \mathcal{B}(X,\mathbb{K})$, where ϱ is defined as in Proposition 1.4.1. Since $(\mathcal{B}(X,\mathbb{K}),\varrho)$ is a complete metric space, $(\mathcal{B}(X,\mathbb{K}), ||\cdot||_{\infty})$ therefore is a Banach algebra. This proves the first claim.

For the second observe that for $f, g \in C_b(X, \mathbb{K})$ and $a \in \mathbb{K}$ the sum f + g, the scalar multiple af, and the product $f \cdot g$ are elements of $C_b(X, \mathbb{K})$ again. To verify this let $x \in X$ and $\varepsilon > 0$. Choose neighborhoods U_1 and U_2 of x so that

$$|f(y) - f(x)| < \min\left\{\frac{\varepsilon}{2}, \frac{\varepsilon}{|a|+1}, \frac{\varepsilon}{2(|g(x)|+1)}\right\} \quad \text{for } y \in U_1$$

and

$$|g(y) - g(x)| < \left\{1, \frac{\varepsilon}{2}, \frac{\varepsilon}{2(|f(x)| + 1)}\right\}$$
 for $y \in U_2$.

Then for all $y \in U_1 \cap U_2$

$$\begin{split} |(f+g)(y) - (f+g)(x)| &\leq |f(y) - f(x)| + |g(y) - g(x)| < \varepsilon ,\\ |(af)(y) - (af)(x)| &\leq |a| \cdot |f(y) - f(x)| < \varepsilon ,\\ |(f \cdot g)(y) - (f \cdot g)(x)| &\leq |g(y)| \cdot |f(y) - f(x)| + |f(x)| \cdot |(g(y) - g(x)| < \varepsilon . \end{split}$$

This means that f + g, af and fg are continuous in x, hence elements of $\mathcal{C}_{b}(X, \mathbb{K})$ since $x \in X$ was arbitrary. So $\mathcal{C}_{b}(X, \mathbb{K})$ is a subalgebra of $\mathcal{B}(X, \mathbb{K})$. By Proposition 1.4.1 one knows that $\mathcal{C}_{b}(X, \mathbb{K})$ is a closed subspace of $\mathcal{B}(X, \mathbb{K})$. The rest of the claim is obvious.

1.4.3 As the next step, we introduce seminorms and their topologies on spaces of differentiable functions defined over an open set $\Omega \subset \mathbb{R}^n$. We agree that from now on Ω will always denote in this section an open subset of \mathbb{R}^n . For any differentiability order $m \in \mathbb{N} \cup \{\infty\}$ the symbol $\mathcal{C}^m(\Omega)$ stands for the space of *m*-times continuously differentiable complex valued functions on Ω . For $i = 1, \ldots, n$ we denote by $x^i : \mathbb{R}^n \to \mathbb{R}$ the *i*-th coordinate function and, if $m \ge 1$, by $\partial_i : \mathcal{C}^m(\Omega) \to \mathcal{C}^{m-1}(\Omega)$ the operator which maps $f \in \mathcal{C}^m(\Omega)$ to the partial derivative $\frac{\partial f}{\partial x^i}$. More generally, if $\alpha \in \mathbb{N}^n$ is a multiindex satisfying $|\alpha| = \alpha_1 + \ldots + \alpha_n \leq m$, then we write $\partial^{\alpha} : \mathcal{C}^m(\Omega) \to \mathcal{C}^{m-|\alpha|}(\Omega)$ for the higher order partial derivative which maps $f \in \mathcal{C}^m(\Omega)$ to $\frac{\partial^{|\alpha|} f}{\partial x_1^{\alpha_1} \cdots \partial x_n^{\alpha_n}}$. Recall that the sum and the product

of two *m*-times differentiable functions and scalar multiples of *m*-times differentiable functions are again *m*-times differentiable, hence $\mathcal{C}^m(\Omega)$ forms a \mathbb{C} -algebra. Now we define $\overline{\mathcal{C}}^m(\Omega)$ to be the space of continuous functions on the closure $\overline{\Omega}$ which are *m*-times continuously differentiable on Ω so that each of its partial derivatives of order $\leq m$ has a continuous extension to $\overline{\Omega}$. Since the operators ∂_i are linear and also derivations by the Leibniz rule, $\overline{\mathcal{C}}^m(\Omega)$ is a subalgebra of $\mathcal{C}^m(\Omega)$. In general, these algebras do not coincide as for example the function $\frac{1}{x}$ on $\mathbb{R}_{>0}$ shows. It is an element of $\overline{\mathcal{C}}^{\infty}(\mathbb{R}_{>0})$ but can not be extended to a continuous function on $\mathbb{R}_{\geq 0}$, so is not an element of $\overline{\mathcal{C}}^{\infty}(\mathbb{R}_{>0})$.

If $X \subset \mathbb{R}^n$ is locally closed which means that X is the intersection of an open and a closed susbet of \mathbb{R}^n , then define $\mathcal{C}^m(X)$ as the quotient space $\mathcal{C}^m(\Omega)/\mathcal{J}_X(\Omega)$, where $\Omega \subset \mathbb{R}^n$ open is chosen so that $X = \overline{X} \cap \Omega$ and where \mathcal{J}_X denotes the ideal sheaf of all *m*-times continuously differentiable functions vanishing on X that is

$$\mathcal{J}_X(\Omega) = \left\{ f \in \mathcal{C}^m(\Omega) \mid f|_X = 0 \right\} \,.$$

Using a smooth partition of unity type of argument one shows that $\mathcal{C}^m(X)$ does not depend on the particular choice of the neighborhood Ω in which X is relatively closed and that $\mathcal{C}^m(X)$ can be naturally identified with the space of continuous functions on X which have an extension to an element of $\mathcal{C}^m(\Omega)$.

1.4.4 Proposition Let $\Omega \subset \mathbb{R}^n$ be open and bounded and $m \in \mathbb{N}_{>0}$. Then $\overline{\mathbb{C}}^m(\Omega)$ equipped with the norm

 $\|\cdot\|_{\Omega,m}:\overline{\mathcal{C}}^m(\Omega)\to\mathbb{R}_{\geq 0},\quad f\mapsto$

A.1.5. Summability

1.5.1 Definition Assume to be given a locally convex topological vector space V over the field K of real or complex numbers. Let $(v_i)_{i\in I}$ be a family of elements of V. Let $\mathcal{F}(I)$ be the set of finite subsets of I and note that it is filtered by set-theoretic inclusion. The family $(v_i)_{i\in I}$ then gives rise to the net $\left(\sum_{i\in J} v_i\right)_{J\in\mathcal{F}(I)}$. One calls the family $(v_i)_{i\in I}$ summable to an element $v \in V$ if the net $\left(\sum_{i\in J} v_i\right)_{J\in\mathcal{F}(I)}$ converges to v. In other words this means that for every convex zero neighborhood

 $U \subset V$ and $\varepsilon > 0$ there exists an element $J_{U,\varepsilon} \in \mathcal{F}(I)$ such that for all finite sets J with $J_{U,\varepsilon} \subset J \subset I$

$$p_U\left(v-\sum_{i\in J}v_i\right)<\varepsilon$$

As before, p_U denotes here the gauge of U. If V is Hausdorff, the limit v of a summable family $(v_i)_{i \in I}$ is uniquely determined, and one writes in this situation

$$v = \sum_{i \in I} v_i \; .$$

We denote the space of summable families in V over the given index set I by $\ell^1(I, V)$. For $E = \mathbb{C}$ we just write $\ell^1(I)$ instead of $\ell^1(I, \mathbb{C})$. If in addition the index set coincides with \mathbb{N} , we briefly denote $\ell^1(\mathbb{N})$ by ℓ^1 .

1.5.2 Proposition (Cauchy criterion for summability) Let V be a complete locally convex topological vector space. A family $(v_i)_{i \in I}$ of elements of V then is summable to some $v \in V$ if and only if it satisfies the following Cauchy condition:

(C) For every convex zero neighborhood $U \subset V$ and $\varepsilon > 0$ there exists an element $J_{U,\varepsilon} \in \mathfrak{F}(I)$ such that for all $K \in \mathfrak{F}(I)$ with $K \cap J_{U,\varepsilon} = \emptyset$ the relation

$$p_U\left(\sum_{i\in K} v_i\right) < \varepsilon$$

holds true.

Proof. By completeness of V it suffices to verify that the net $\left(\sum_{i\in J} v_i\right)_{J\in\mathcal{F}(I)}$ is a Cauchy net if and only if condition (C) is satisfied. Recall that one calls $\left(\sum_{i\in J} v_i\right)_{J\in\mathcal{F}(I)}$ a Cauchy net if for every convex zero neighborhood $U \subset V$ all $\varepsilon > 0$ there exists an element $J_{U,\varepsilon} \in \mathcal{F}(I)$ such that for all $J, J' \in \mathcal{F}(I)$ containing $J_{U,\varepsilon}$ as a subset the relation

$$p_U\left(\sum_{i\in J} v_i - \sum_{i\in J'} v_i\right) < \varepsilon$$

holds true. But that is clearly equivalent to condition (C).

1.5.3 Several other notions of summability have been introduced in the analysis and functional analysis literature. These are mainly either used to establish summability criteria or are used in the study of topological tensor products and nuclearity of locally convex topological vector spaces, see Grothendieck (1955); Pietsch (1972). In the following we define these further notions of summability and study their properties. The symbol V hereby always stands for a locally convex tvs, I always denotes a nonempty index set, and $\mathcal{F}(I)$ the set of its finite subsets.

1.5.4 Definition A family $(v_i)_{i \in I}$ in V is called *weakly summable* to $v \in V$ if for every continuous linear form $\alpha : V \to \mathbb{K}$ the net $\left(\sum_{i \in J} \alpha(v_i)\right)_{J \in \mathcal{F}(I)}$ converges in \mathbb{K} to $\alpha(v)$. In other words this means that for every $\alpha \in V'$ and $\varepsilon > 0$ there exists a finite set $J_{\alpha,\varepsilon} \subset I$ such that for all finite sets J with $J_{\alpha,\varepsilon} \subset J \subset I$

$$\left|\alpha(v) - \sum_{j \in J} \alpha(v_i)\right| < \varepsilon$$

The set of all weakly summable families in V with index set I is denoted $\ell^1[I, V]$.

1.5.5 Definition A family $(v_i)_{i \in I}$ in V is called *absolutely summable* if for every circled convex zero neighborhood $U \subset V$ there exists some $C \ge 0$ such that

$$\sum_{i\in J}p_{U}\left(v_{i}\right)\leqslant C\quad\text{for all }J\in\mathcal{F}(I)\ .$$

We denote the set of all absolutely summable families in V by $\ell^1\{I, V\}$.

1.5.6 Proposition A family $(v_i)_{i \in I} \subset V$ is absolutely summable if and only if for every element U of a basis of circled convex zero neighborhoods there exists a $C \ge 0$ such that

$$\sum_{i \in J} p_U(v_i) \leqslant C \quad \text{for all } J \in \mathcal{F}(I) \ .$$

Proof.

1.5.7 Definition A family $(v_i)_{i \in I}$ in V is called *totally summable* if there exists a bounded absolutely convex subset $B \subset V$ and a $C \ge 0$ such that

$$\sum_{i \in J} p_B\left(v_i\right) \leqslant C \quad \text{for all } J \in \mathcal{F}(I) \ .$$

We write $\ell^1 \langle I, V \rangle$ for the set of all totally summable families in V.

Summable families of complex numbers

1.5.8 Lemma (cf. (Pietsch, 1972, Lem. 1.1.2)) Let $(z_i)_{i \in I}$ be a family of complex numbers for which there exists a positive real number C > 0 such that

$$\left|\sum_{i\in J} z_i\right| \leqslant C \quad \text{for all } J \in \mathcal{F}(I) \ .$$

Then one has the estimate

$$\sum_{i\in J} |z_i| \leqslant 4C \quad \text{for all } J\in \mathfrak{F}(I) \,\,.$$

Proof. We assume first that all z_i are real. Then let I^+ the set of all indices $i \in I$ such that $z_i \ge 0$, and I^- the set of all $i \in I$ such that $z_i < 0$. Then, for all finite $J \subset I$

$$\sum_{i\in J} |z_i| = \sum_{i\in J\cap I^+} |z_i| + \sum_{i\in J\cap I^-} |z_i| = \left|\sum_{i\in J\cap I^+} z_i\right| + \left|\sum_{i\in J\cap I^-} z_i\right| \leqslant 2C \ .$$

In the general case decompose z_i into real and imaginary parts $x_i = \Re \mathfrak{e} z_i$ and $y_i = \Im \mathfrak{m} z_i$. By the triangle inequality one obtains for all finite $J \subset I$

$$\sum_{i \in J} |z_i| \leq \sum_{i \in J} |x_i| + \sum_{i \in J} |y_i| \leq 4C .$$

1.5.9 Proposition For a family $(z_i)_{i \in I}$ of complex numbers the following are equivalent.

- (i) The family $(z_i)_{i \in I}$ is summable.
- (ii) The family $(|z_i|)_{i \in I}$ is summable.
- (iii) The family $(z_i)_{i \in I}$ is absolutely summable.
- (iv) There exists some C > 0 such that $\sum_{i \in J} |z_i| \leq C$ for all $J \in \mathfrak{F}(I)$.

In case that one hence all of the conditions are fulfilled, the estimate

$$\left|\sum_{i\in I} z_i\right| \leqslant \sum_{i\in I} |z_i|$$

holds true.

Proof. Assume that $(z_i)_{i \in I}$ is absolutely summable. Since \mathbb{C} is normed with norm given by the absolut value this just means that there exists some C > 0 such that $\sum_{i \in J} |z_i| \leq C$ for all $J \in \mathcal{F}(I)$. Hence the supremum $c = \sup \{\sum_{i \in J} |z_i| \mid J \in \mathcal{F}(I)\}$ exists and is $\leq C$. For given $\varepsilon > 0$ choose $J_{\varepsilon} \in \mathcal{F}(I)$ such that

$$c - \varepsilon \leq \sum_{i \in J_{\varepsilon}} |z_i| \leq c$$
.

Then one has for all $K \in \mathcal{F}(I)$ with $K \cap J_{\varepsilon} = \emptyset$

$$\left|\sum_{i\in K} z_i\right| \leq \sum_{i\in K} |z_i| \leq \varepsilon \; .$$

Hence $(\sum_{i \in J} z_i)_{J \in \mathcal{F}(I)}$ is a Cauchy net, so has to converges by completeness of \mathbb{C} . This proves summability of $(z_i)_{i \in I}$.

Vice versa, assume now that $(z_i)_{i \in I}$ is summable. Then $(\sum_{i \in J} z_i)_{J \in \mathcal{F}(I)}$ is a Cauchy net. Hence there exists an element $J_1 \in \mathcal{F}(I)$ such that for all $K \in \mathcal{F}(I)$ with $K \cap J_1 = \emptyset$ the inequality

$$\left|\sum_{i\in K} z_i\right| < 1$$

holds true. Let $C = \sum_{i \in J_1} |z_i|$. Then one has for all $J \in \mathcal{F}(I)$

$$\left|\sum_{i\in J} z_i\right| \leqslant \left|\sum_{i\in J\setminus J_1} z_i\right| + \left|\sum_{i\in J\cap J_1} z_i\right| \leqslant 1 + C.$$

By the preceding lemma the set of partial sums $\sum_{i \in J} |z_i|$, where J runs through the finite subsets of I, is then bounded by 4 + 4C, hence $(z_i)_{i \in I}$ is absolutely summable.

Summability in Banach spaces

1.5.10 Proposition Let V be a normed vector space. For a family $(v_i)_{i \in I}$ of elements in V the following are equivalent:

- (i) The family $(v_i)_{i \in I}$ is absolutely summable.
- (ii) The family $(||v_i||)_{i \in I}$ is summable.
- (iii) There exists some C > 0 such that $\sum_{i \in J} ||v_i|| \leq C$ for all $J \in \mathcal{F}(I)$.
- If V is even a Banach space, these conditions are all equivalent to

(iv) The family $(v_i)_{i \in I}$ is summable.

Proof. (ii) and (iii) are equivalent by Proposition 1.5.9 Assume now that (i) holds true.

to do: Carl Neumann series

Properties of and relations between the various summability types

1.5.11 Theorem Let I be a non-empty index set. Then the spaces $\ell^1(I, V)$ of summable families, $\ell^1[I, V]$ of weakly summable families, $\ell^1\{I, V\}$ of absolutely summable families and $\ell^1\langle I, V\rangle$ of totally summable families in E are all subvector spaces of the product vector space $E^I = \prod_{i \in I} E$. Furthermore one has the following chain of inclusions:

$$\ell^1 \langle I, V \rangle \subset \ell^1 \{I, V\}$$
 and $\ell^1 (I, V) \subset \ell^1 [I, V]$.

If E is complete, then one even has

$$\ell^1\{I, \mathcal{V}\} \subset \ell^1(I, \mathcal{V})$$

Proof. Now let (v_i) be a summable family and $\alpha : V \to \mathbb{K}$ a continuous linear form.

Let U be an absolutely convex zero neighborhood. Then U absorbes B, so there exists r > 0 such that $B \subset rU$. Hence

A.1.6. Topological tensor products

1.6.1 Definition (cf. (Grothendieck, 1955, Chap. I, § 3, n° 3)) Let V and W be two locally convex topological vector spaces over the ground field \mathbb{K} . A locally convex vector topology τ on the (algebraic) tensor product $V \otimes W$ is called *compatible with the tensor product structure*, an *admissible tensor product topology* or just *admissible* if the following conditions hold true:

(ATPT1) The canonical map $V \times W \rightarrow V \otimes_{\tau} W$ is separately continuous that is for each $v \in V$ and each $w \in W$ the linear maps

$$W \to V \otimes_{\tau} W, y \mapsto v \otimes y \text{ and } V \to V \otimes_{\tau} W, x \mapsto x \otimes w$$

are continuous where $V \otimes_{\tau} W$ denotes the vector space $V \otimes W$ equipped with τ .

- (ATPT2) For all linear maps $\alpha \in V'$ and $\beta \in W'$ the canonical linear map map $\alpha \otimes \beta : V \otimes_{\tau} W \to \mathbb{K}$ is continuous.
- (ATPT3) For every equicontinuous subset $A \subset V'$ and equicontinuous subset $B \subset W'$ the set $\{\alpha \otimes \beta \mid \alpha \in A \& \beta \in B\}$ is an equicontinuous subset of the topological dual of $V \otimes_{\tau} W$.

The locally convex vector topology τ is called *strongly compatible with the tensor product structure*, a *strongly admissible tensor product topology* or briefly *strongly admissible* if it satisfies:

(sATPT) The canonical map $V \times W \rightarrow V \otimes_{\tau} W$ is continuous where $V \times W$ carries the product topology.

1.6.2 The admissible respectively strongly admissible vector topologies on $\rm V\otimes W$ are obviously partially ordered by set-theoretic inclusion. Therefore, the following definition makes sense.

1.6.3 Definition

A.2. Distributions and Fourier Transform

A.2.1. Schwartz distributions

A.2.2. Pullback of distributions

2.2.1 Let M and N be smooth manifolds and $f: M \to N$ a smooth map. One then has a continuous pullback map $f^*: \mathcal{C}^{\infty}(N) \to \mathcal{C}^{\infty}(M)$ which maps an element $h \in \mathcal{C}^{\infty}(N)$ to the composition $h \circ f: M \to \mathbb{R}$ which obviously is a smooth function on M. The Faà-di-Bruno formula from Theorem 8.1.10 tells that f^* is continuous indeed. In this section we want to establish criteria under which the pullback of functions can be extended to a pullback of distributions. We also will study continuity properties of the distributional pullback operation

Let us start with the following observation.

2.2.2 Lemma Let $f: U_1 \to U_2$ be a diffeomorphism between two open subsets $U_1, U_2 \subset \mathbb{R}^n$ and λ the Lebesgue measure on \mathbb{R}^n . Then for every $u \in \mathcal{C}(U_2)$ and $\varphi \in \mathcal{D}(U_1)$ the equality

$$\int_{U_1} \varphi f^* u \, d\lambda = \int_{U_2} (\varphi \circ f^{-1}) \, u \left| \det D f^{-1} \right| d\lambda$$

holds true.

Proof. The claim is an immediate consequence of the change-of-variables formula.

2.2.3 Using the lemma as guideline we now extend the pullback of functions to distributions. Denote for $U \subset \mathbb{R}^n$ by $\langle \cdot, \cdot \rangle$ the pairing between $\mathcal{D}'(U)$ and $\mathcal{D}(U)$. Under the assumptions of the lemma assume u to be a distribution on U_1 that is an element of $\mathcal{D}'(U_2)$. Then the map

$$f^*u: \mathcal{D}(U_1) \to \mathbb{R}, \quad \varphi \mapsto \langle u, |\det Df^{-1}| (f^{-1})^* \varphi \rangle.$$

is an element of the distribution space $\mathcal{D}'(U_1)$ since the map

$$\mathcal{D}(U_1) \to \mathcal{D}(U_2), \quad \varphi \mapsto |\det Df^{-1}| \varphi \circ f^{-1}$$

is linear and continuous with respect to the LF-topologies on $\mathcal{D}(U_1)$ and $\mathcal{D}(U_2)$. One calls f^*u the *pullback* of the distribution u under f. By Lemma 2.2.2, this pullback operation extends the one for continuous functions and it is obviously uniquely determined by that property.

We continue with another observation.

2.2.4 Lemma Assume that $U \subset \mathbb{R}^n$ is open, $f : U \to \mathbb{R}$ a submersion and $\varphi \in \mathcal{D}(U)$ a test function. Then the map

$$f_*\varphi: \mathbb{R} \to \mathbb{R}, \ t \mapsto \frac{d}{dt} \int_{\{x \in U \mid f(x) < t\}} \varphi(x) dx$$

is well-defined, smooth and has compact support.

Proof. Let us assume first that the map $\Psi: U \to \mathbb{R}^n$, $(x_1, \ldots, x_n) \mapsto ()$ is \Box

todo Possibly assume that M is orientable and carries a volume form.

A.2.3. Hyperfunctions of a single variable

2.3.1 Let us introduce some notation. For every open intervall $I \subset \mathbb{R}$ call an open subset $U \subset \mathbb{C}$ such that $I = U \cap \mathbb{R}$ a *complex neighborhood* of I. Denote by \mathbb{C}^+ the upper complex half-plane $\{z \in \mathbb{C} \mid \Im m z > 0\}$ and by \mathbb{C}^- the lower complex half-plane $\{z \in \mathbb{C} \mid \Im m z < 0\}$. More generally, put $U^+ = U \cap \mathbb{C}^+$ and $U^- = U \cap \mathbb{C}^-$ for every open subset $U \subset \mathbb{C}$.

A.3. Hilbert Spaces

A.3.1. Inner product spaces

3.1.1 Let us first remind the reader that as before \mathbb{K} stands for the field of real or of complex numbers. We will keep this notational agreement throughout the whole chapter.

3.1.2 Definition By a *sesquilinear form* on a \mathbb{K} -vector space V one understands a map $\langle \cdot, \cdot \rangle$: $V \times V \to \mathbb{K}$ with the following two properties:

(SF1) The map $\langle \cdot, \cdot \rangle$ is *conjugate-linear* in its first coordinate which means that

$$\langle v_1 + v_2, w \rangle = \langle v_1, w \rangle + \langle v_2, w \rangle$$
 and $\langle rv, w \rangle = \overline{r} \langle v, w \rangle$

for all $v, v_1, v_2, w \in V$ and $r \in \mathbb{K}$.

(SF2) The map $\langle \cdot, \cdot \rangle$ is *linear* in its second coordinate which means that

$$\langle v, w_1 + w_2 \rangle = \langle v, w_1 \rangle + \langle v, w_2 \rangle$$
 and $\langle v, rw \rangle = r \langle v, w \rangle$

for all $v, w, w_1, w_2 \in V$ and $r \in \mathbb{K}$.

A hermitian form is a sesquilinear form $\langle \cdot, \cdot \rangle$ on V with the following additional property:

(SF3) The map $\langle \cdot, \cdot \rangle$ is *conjugate-symmetric* which means that

$$\langle v, w \rangle = \overline{\langle w, v \rangle}$$
 for all $v, w \in V$.

A sesquilinear form $\langle \cdot, \cdot \rangle$ is called *weakly-nondegenerate* if it satisfies axiom

(SF4w) For every $v \in V$, the map $V \to \mathbb{K}$, $w \to \langle w, v \rangle$ is the zero map if and only if v = 0.

Finally, one calls a hermitian form $\langle \cdot, \cdot \rangle$ on V positive semidefinite if

(SF5s) $\langle v, v \rangle \ge 0$ for all $v \in V$.

3.1.3 Remark Recall that a map $\langle \cdot, \cdot \rangle : V \times V \to \mathbb{K}$ is called *bilinear* if it satisfies (SF2) and the following condition:

(BF1) The map $\langle \cdot, \cdot \rangle$ is *linear* in its first coordinate which means that

$$\langle v_1 + v_2, w \rangle = \langle v_1, w \rangle + \langle v_2, w \rangle$$
 and $\langle rv, w \rangle = r \langle v, w \rangle$

for all $v, v_1, v_2, w \in V$ and $r \in \mathbb{K}$.

In case the underlying ground field \mathbb{K} coincides with the field of real numbers, a sesquilinear form is by definition the same as a bilinear form, and a hermitian form the same as a symmetric bilinear form.

3.1.4 Given a positive semidefinite hermitian form $\langle \cdot, \cdot \rangle$ on a K-vector space V, one calls two vectors $v, w \in V$ orthogonal if $\langle v, w \rangle = 0$. Since the hermitian form $\langle \cdot, \cdot \rangle$ is assumed to be positive semidefinite, the map

$$\|\cdot\|: \mathcal{V} \to \mathbb{R}_{\geq 0}, \ v \mapsto \|v\| = \sqrt{\langle v, v \rangle}$$

is well-defined. We will later see that $\|\cdot\|$ is a seminorm on V and therefore call the map $\|\cdot\|$ the seminorm associated to $\langle\cdot,\cdot\rangle$. The following formulas are immediate consequences of the properties defining a positive semidefinite hermitian form and the definition of the associated seminorm:

$$\|v+w\|^{2} = \|v\|^{2} + 2\Re(\langle v,w\rangle + \|w\|^{2} \text{ for all } v,w \in V , \qquad (A.3.1.1)$$

$$\|v+w\|^2 = \|v\|^2 + \|w\|^2$$
 for all orthogonal $v, w \in V$, (A.3.1.2)

$$\|v+w\|^2 + \|v-w\|^2 = 2(\|v\|^2 + \|w\|^2) \text{ for all } v, w \in V , \qquad (A.3.1.3)$$

$$\|rv\| = \sqrt{|r|^2 \langle v, v \rangle} = |r| \|v\|$$
 for all $v, w \in V$ and $r \in \mathbb{K}$. (A.3.1.4)

Formula (A.3.1.2) is an abstract version of the *pythagorean theorem*, Equation (A.3.1.3) is called the *parallelogram identity*. The triangle inequality for the map $\|\cdot\|$ will turn out to be a consequence of the next result.

3.1.5 Proposition (Cauchy–Schwarz inequality) Given a positive semidefinite hermitian form $\langle \cdot, \cdot \rangle$ on a \mathbb{K} -vector space V the following inequality holds true:

$$|\langle v, w \rangle| \le ||v|| ||w|| \quad \text{for all } v, w \in \mathcal{V}. \tag{A.3.1.5}$$

Equality holds if v and w are linearly dependent. In case $\langle \cdot, \cdot \rangle$ is positive definite, the converse holds true as well.

Proof. First consider the case where ||v|| = ||w|| = 0. Note that this does not imply that v = 0 (or w = 0) unless the hermitian form $\langle \cdot, \cdot \rangle$ is positive definite. Now put $c = -\langle v, w \rangle$ and compute

$$0 \leq \|cv + w\|^2 = 2 \operatorname{\mathfrak{Re}}(\bar{c}\langle v, w \rangle) = -2|\langle v, w \rangle|^2 .$$
(A.3.1.6)

This entails $\langle v, w \rangle = 0$ and the Cauchy–Schwarz inequality is proved for ||v|| = ||w|| = 0.

If $||v|| \neq 0$ or $||w|| \neq 0$, we can assume without loss of generality that $||v|| \neq 0$. Under this assumption put

$$c = -\frac{\langle v, w \rangle}{\|v\|^2}$$

and compute

$$0 \leq \|cv + w\|^{2} = |c|^{2} \|v\|^{2} + 2 \Re(\bar{c}\langle v, w \rangle) + \|w\|^{2} = = \frac{|\langle v, w \rangle|^{2}}{\|v\|^{2}} - 2 \frac{|\langle v, w \rangle|^{2}}{\|v\|^{2}} + \|w\|^{2} = \|w\|^{2} - \frac{|\langle v, w \rangle|^{2}}{\|v\|^{2}} .$$
(A.3.1.7)

Hence the estimate

$$|\langle v, w \rangle|^2 \leqslant \|v\|^2 \|w\|^2$$

holds which entails the Cauchy-Schwarz inequality.

In the case where v, w are linearly dependant nonzero elements of V there exists a nonzero scalar $a \in \mathbb{K}$ such that v = aw. Therefore

$$|\langle v, w \rangle| = |a| ||w||^2 = ||v|| ||w||.$$

If one of v or w is 0, then both sides of the Cauchy–Schwarz inequality are 0.

In the positive definite case, equality in (A.3.1.5) entails by Equation (A.3.1.7) that cv + w = 0 whenever $v \neq 0$. If v = 0, then $v = 0 \cdot w$. In either case this means that v and w are linearly dependent.

3.1.6 Lemma A positive semidefinite hermitian form $\langle \cdot, \cdot \rangle$ on a \mathbb{K} -vector space V is weakly-nondegenerate if and only if it is positive definite that is if and only if

(SF5p) $\langle v, v \rangle > 0$ for all $v \in V \setminus \{0\}$.

Proof. A positive definite real bilinear or complex hermitian form $\langle \cdot, \cdot \rangle$ is weakly-nondegenerate since for every $v \in V \setminus \{0\}$ the linear form $\langle v, - \rangle : V \to \mathbb{K}$ is nonzero by $\langle v, v \rangle > 0$.

Conversely, if $\langle v, - \rangle : V \to \mathbb{K}$ is nonzero for all $v \in V \setminus \{0\}$, then there exists an element $w \in V$ such that $\langle w, v \rangle \neq 0$. The Cauchy–Schwarz inequality entails

$$0 < |\langle w, v \rangle|^2 \le \langle w, w \rangle \langle v, v \rangle ,$$

which implies $\langle v, v \rangle > 0$. Hence $\langle \cdot, \cdot \rangle$ is positive definite.

3.1.7 Proposition The map

$$\|\cdot\|: V \to \mathbb{R}_{\geq 0}, v \mapsto \|v\| = \sqrt{\langle v, v \rangle}$$

associated to a positive semidefinite hermitian form $\langle \cdot, \cdot \rangle$ on a \mathbb{K} -vector space V is a seminorm. If the hermitian form is positive definite, then $\|\cdot\|$ is even a norm.

Proof. Absolute homogeneity (N1) is given by Eq. (A.3.1.4). The triangle inequality is a consequence of the Cauchy–Schwarz inequality:

$$\|v+w\|^{2} = \|v\|^{2} + 2 \Re \mathfrak{e} \langle v, w \rangle + \|w\|^{2} \leq \|v\|^{2} + 2 \|v\| \|w\| + \|w\|^{2} = \left(\|v\| + \|w\|\right)^{2}$$

 $\text{Finally, if } \langle \cdot, \cdot \rangle \text{ is positive definite, then } \|v\| = \sqrt{\langle v, v \rangle} > 0 \text{ for all } v \in \mathrm{V} \setminus \{0\} \text{, so } \|\cdot\| \text{ is a norm.} \quad \Box$

3.1.8 Definition By an *inner product* or a *scalar product* on a \mathbb{K} -vector space \mathcal{H} one understands a positive definite hermitian form on \mathcal{H} . A \mathbb{K} -vector space \mathcal{H} endowed with an inner product $\langle \cdot, \cdot \rangle : \mathcal{H} \times \mathcal{H} \to \mathbb{K}$ is called an *inner product space* or a *pre-Hilbert space*.

A hermitian form on a \mathbb{K} -vector space \mathcal{H} which is only positive semidefinite is called a *semi-inner* product or a *semi-scalar product*.

A *Hilbert space* is an inner product space $(\mathcal{H}, \langle \cdot, \cdot \rangle)$ which is complete as a normed vector space. In other words, a Hilbert space is Banach space where the norm on the space is induced by an inner product.

3.1.9 Examples (a) The vector space \mathbb{R}^n with the *euclidean inner product*

$$\langle \cdot, \cdot \rangle : \mathbb{R}^n \times \mathbb{R}^n \to \mathbb{R}, \ ((v_1, \dots, v_n), (w_1, \dots, w_n)) \mapsto \sum_{i=1}^n v_i w_i$$

is a real Hilbert space. Obviously, $\langle \cdot, \cdot \rangle$ is linear in the first argument, symmetric, and positive definite, hence a real inner product. The associated norm is the *euclidean norm*. We have seen before that \mathbb{R}^n with the euclidean norm is complete.

(b) The vector space \mathbb{C}^n together with the hermitian form

$$\langle \cdot, \cdot \rangle : \mathbb{C}^n \times \mathbb{C}^n \to \mathbb{C}, \ ((v_1, \dots, v_n), (w_1, \dots, w_n)) \mapsto \sum_{i=1}^n \overline{v}_i w_i$$

is a complex Hilbert space. One immediately verifies that $\langle \cdot, \cdot \rangle$ is linear in the second argument, conjugate-symmetric, and positive definite. Hence $\langle \cdot, \cdot \rangle$ is a complex inner product which we sometimes call the *standard hermitian inner product* on \mathbb{C}^n . Its associated norm is again the euclidean norm, so by completeness of $\mathbb{C}^n \cong \mathbb{R}^{2n}$ with respect to the euclidean norm one obtains the claim.

(c) The set

$$\ell^2 = \left\{ (z_k)_{k \in \mathbb{N}} \in \mathbb{C}^{\mathbb{N}} \mid \sum_{k=0}^{\infty} |z_k|^2 < \infty \right\}$$

of square summable sequences of complex numbers is a complex Hilbert space with inner product

$$\langle \cdot, \cdot \rangle : \ell^2 \times \ell^2 \to \mathbb{C}, \ \left((z_k)_{k \in \mathbb{N}}, (w_k)_{k \in \mathbb{N}} \right) \mapsto \sum_{k=0}^{\infty} \overline{z}_k w_k$$

To prove this one needs to first verify that ℓ^2 is a subvector space of $\mathbb{C}^{\mathbb{N}}$. For $z = (z_k)_{k \in \mathbb{N}} \in \mathbb{C}^{\mathbb{N}}$ denote by ||z|| the *extended norm* $\sqrt{\sum_{k=0}^{\infty} |z_k|^2} = \sup_{K \in \mathbb{N}} \sqrt{\sum_{k=0}^{K} |z_k|^2} \in [0, \infty]$. Then $z \in \ell^2$ if and only if $||z|| < \infty$. Now let $a \in \mathbb{C}$ and $z \in \ell^2$ and compute

$$||az|| = \sqrt{\sum_{k=0}^{\infty} |az_k|^2} = |a| \sqrt{\sum_{k=0}^{\infty} |z_k|^2} = |a| \cdot ||z|| < \infty .$$

Hence $az \in \ell^2$. If $z, w \in \ell^2$, denote for each $K \in \mathbb{N}$ by $z_{(K)}$ and $w_{(K)}$ the "cut-off" vectors $(z_0, \ldots, z_K) \in \mathbb{C}^{K+1}$ and $(w_0, \ldots, w_K) \in \mathbb{C}^{K+1}$, respectively. By the triangle inequality for the norm on the Hilbert space \mathbb{C}^{K+1} one concludes

$$\sqrt{\sum_{k=0}^{K} |z_k + w_k|^2} = ||z_{(K)} + w_{(K)}|| \le ||z_{(K)}|| + ||w_{(K)}|| \le ||z|| + ||w|| < \infty$$

Therefore, the sequence of partial sums $\sum_{k=0}^{K} |z_k + w_k|^2$, $K \in \mathbb{N}$, is bounded, so convergent by the the monotone convergence theorem. One obtains

$$||z + w|| = \lim_{K \to \infty} \sqrt{\sum_{k=0}^{K} |z_k + w_k|^2} \le ||z|| + ||w|| < \infty$$
.

Hence z + w is square summable and ℓ^2 a vector subspace of $\mathbb{C}^{\mathbb{N}}$ indeed. Note that our argument also shows that the restriction of the extended norm to ℓ^2 is a norm.

We need to show that $\langle \cdot, \cdot \rangle$ is well-defined. To this end it suffices to prove that for all $z, w \in \ell^2$ the family $(z_k \overline{w}_k)_{k \in \mathbb{N}}$ is absolutely summable or in other words that $\sum_{k=0}^{\infty} |z_k \overline{w}_k| < \infty$. One concludes by the Hölder inequality for sums

$$\sum_{k=0}^{K} |\overline{z}_k w_k| = \sum_{k=0}^{K} |z_k w_k| \le ||z_{(K)}|| ||w_{(K)}|| \le ||z|| ||w|| .$$

So the left hand side has an upper bound uniform in K which by the monotone convergence theorem entails convergence of the partial sums and the estimate

$$\sum_{k=0}^{\infty} |\overline{z}_k w_k| \leqslant ||z|| ||w|| < \infty .$$

By definition it is clear that $\langle \cdot, \cdot \rangle$ is linear in the second argument, conjugate-symmetric and positive definite, hence a complex inner product. Note that the norm associated to $\langle \cdot, \cdot \rangle$ coincides with the above defined map $\|\cdot\|$.

It remains to be shown that ℓ^2 is complete. Let $(z^n)_{n\in\mathbb{N}}$ with $z^n = (z_k^n)_{k\in\mathbb{N}} \in \ell^2$ for all $n\in\mathbb{N}$ be a Cauchy sequence in ℓ^2 . For $\varepsilon > 0$ choose $N_{\varepsilon} \in \mathbb{N}$ so that

$$||z^n - z^m|| < \varepsilon$$
 for all $n, m \ge N_{\varepsilon}$.

For each fixed $k \in \mathbb{N}$ one therefore has

$$|z_k^n - z_k^m| \le ||z^n - z^m|| < \varepsilon \quad \text{for all } n, m \ge N_\varepsilon . \tag{A.3.1.8}$$

By completeness of \mathbb{C} there exist $z_k \in \mathbb{C}$ such that $\lim_{n \to \infty} z_k^n = z_k$ for all $k \in \mathbb{N}$. We claim that $z = (z_k)_{k \in \mathbb{N}}$ is an element of ℓ^2 and that $(z^n)_{n \in \mathbb{N}}$ converges to z. To verify this observe that for all $\varepsilon > 0$, $K \in \mathbb{N}$ and $n \ge N_{\varepsilon}$

$$\sum_{k=0}^{K} |z_k - z_k^n|^2 = \lim_{m \to \infty} \sum_{k=0}^{K} |z_k^m - z_k^n|^2 \leq \sup_{m \ge N_{\varepsilon}} \sum_{k=0}^{K} |z_k^m - z_k^n|^2 \leq \sup_{m \ge N_{\varepsilon}} \|z^m - z^n\|^2 \leq \varepsilon^2 .$$

This implies by the triangle inequality and the fact that the Cauchy sequence $(z^n)_{n\in\mathbb{N}}$ is bounded in norm by some C > 0 that for all $K \in \mathbb{N}$ and $N = N_1$

$$\sqrt{\sum_{k=0}^{K} |z_k|^2} = ||z_{(K)}|| \le ||z_{(K)} - z_{(K)}^N|| + ||z_{(K)}^N|| \le ||z_{(K)} - z_{(K)}^N|| + ||z^N|| \le 1 + C.$$

Hence $||z|| = \sqrt{\sum_{k=0}^{\infty} |z_k|^2} \leqslant 1 + C$ and $z \in \ell^2$. In addition one obtains

$$||z - z^n|| = \lim_{K \to \infty} \sqrt{\sum_{k=0}^K |z_k - z_k^n|^2} \leqslant \varepsilon \quad \text{for all } n \ge N_{\varepsilon} \;.$$

This means that z is the limit of the sequence $(z^n)_{n\in\mathbb{N}}$ and ℓ^2 is complete.

(d) Denote by λ the Lebesgue measure and let

$$\mathcal{L}^2(\mathbb{R}^d) = \left\{ f: \mathbb{R}^d \to \mathbb{C} \; \middle| \; f \text{ is Lebesgue measurable and } \|f\|_2 := \sqrt{\int_{\mathbb{R}^d} |f|^2 d\lambda} < \infty \right\}$$

be the space of Lebesgue square integrable functions on \mathbb{R}^d . Then $\mathcal{L}^2(\mathbb{R}^d)$ is a linear subspace of the space of all measurable functions by Minkowski's inequality which reads

$$\|f + g\|_p \leq \|f\|_p + \|g\|_p$$
 for all measurable $f, g: \mathbb{R}^d \to \mathbb{C}$.

Hereby, $||f||_p$ denotes for $p \in [1, \infty)$ the \mathcal{L}^p -seminorm $(\int_{\mathbb{R}^d} |f|^p d\lambda)^{1/p}$ of a measurable function $f : \mathbb{R}^d \to \mathbb{C}$. Note that $||f||_p$ can attain the value ∞ , namely when f is not in the space $\mathcal{L}^p(\mathbb{R}^d)$. By Hölder's inequality, the product fg is Lebesgue integrable for $f, g \in \mathcal{L}^2(\mathbb{R}^d)$ and one has the estimate

$$\int_{\mathbb{R}^d} |fg| d\lambda = \|fg\|_1 \leqslant \|f\|_2 \, \|g\|_2 \, \, .$$

Hence the map

$$\langle \cdot, \cdot \rangle : \mathcal{L}^2(\mathbb{R}^d) \times \mathcal{L}^2(\mathbb{R}^d) \to \mathbb{C}, \ (f,g) \mapsto \int_{\mathbb{R}^d} \overline{f}g \, d\lambda$$

is well-defined and a positive semidefinite hermitian form on $\mathcal{L}^2(\mathbb{R}^d)$. By construction, the associated seminorm is the \mathcal{L}^2 -seminorm $\|\cdot\|_2$. Modding out $\mathcal{L}^2(\mathbb{R}^d)$ by the kernel

$$\mathcal{N} := \operatorname{Ker}(\|\cdot\|_2) = \left\{ f \in \mathcal{L}^2(\mathbb{R}^d) \ \bigg| \ \int_{\mathbb{R}^d} |f|^2 d\lambda = 0 \right\}$$

gives the Lebesgue space

$$L^2(\mathbb{R}^d) := \mathcal{L}^2(\mathbb{R}^d) / \mathcal{N}$$
.

The hermitian form $\langle \cdot, \cdot \rangle$ vanishes on $\mathcal{N} \times \mathcal{L}^2(\mathbb{R}^d)$ and $\mathcal{L}^2(\mathbb{R}^d) \times \mathcal{N}$ by the Cauchy–Schwarz inequality, hence descends to a hermitian form

$$\langle \cdot, \cdot \rangle : L^2(\mathbb{R}^d) \times L^2(\mathbb{R}^d) \to \mathbb{C}, \ (f + \mathcal{N}, g + \mathcal{N}) \mapsto \int_{\mathbb{R}^d} \overline{f}g \, d\lambda \ .$$

That hermitian form is positive definite, since $\langle f + \mathcal{N}, f + \mathcal{N} \rangle = 0$ means $\int_{\mathbb{R}^d} |f|^2 d\lambda = 0$, hence $f \in \mathcal{N}$. Let us show that $L^2(\mathbb{R}^d)$ is complete with respect to the L^2 -norm $\|\cdot\|_2$ induced by the inner product. Note that on the quotient space $\|\cdot\|_2$ is a norm indeed by construction. So let $(f_n + \mathcal{N})_{n \in \mathbb{N}}$ be a Cauchy sequence in $L^2(\mathbb{R}^d)$. Choose a subsequence $(f_{n_k})_{k \in \mathbb{N}}$ such that

$$\|f_{n_k} - f_{n_{k-1}}\|_2 < \frac{1}{2^k}$$
 for all $k \in \mathbb{N}_{>0}$

and put

$$g_n(x) = \sum_{k=1}^n |f_{n_k}(x) - f_{n_{k-1}}(x)|$$
 for $x \in \mathbb{R}^d$ and $n \in \mathbb{N}$.

The limit function

$$g: \mathbb{R}^d \to [0, \infty], x \mapsto \lim_{n \to \infty} g_n(x) = \liminf_{n \to \infty} g_n(x)$$

then exists even though it might not be finite everyhwere. Minkowski's inequality for the \mathcal{L}^2 -norm entails that $||g_n||_2 \leq 1$ for all $n \in \mathbb{N}$, hence g is measurable and $||g||_2 \leq \liminf_{n \to \infty} ||g_n||_2 \leq 1$ by Fatou's lemma. Therefore, g(x) is finite for all x up to a set $Z \subset \mathbb{R}^d$ of measure 0, and for those xthe series with partial sums $g_n(x)$ converges absolutely. For all $x \in \mathbb{R}^d \setminus Z$ the limit

$$f(x) = \lim_{k \to \infty} f_{n_k}(x) = f_{n_0} + \lim_{k \to \infty} \sum_{j=1}^k (f_{n_j}(x) - f_{n_{j-1}}(x))$$

therefore exists in \mathbb{C} . Put f(x) = 0 for all $x \in Z$, and let $\chi_Z : \mathbb{R}^d \to \mathbb{R}$ be the characteristic function of Z. Then the sequence of functions $(\chi_Z f_{n_k})_{k \in \mathbb{N}}$ converges pointwise to f, and each of the functions $\chi_Z f_n$ is measurable, actually even square integrable. Since

$$|\chi_Z f_{n_k}| \leq |\chi_Z f_{n_0}| + g_k \leq |\chi_Z f_{n_0}| + g \quad \text{for all } k \in \mathbb{N}$$

and since $|\chi_Z f_{n_0}| + g$ is square integrable by Minkowski's inequality, the pointwise limit f is square integrable by Lebesgue's dominated convergence theorem, and $f + \mathcal{N}$ is in $L^2(\mathbb{R}^d)$. It remains to show that $(f_n + \mathcal{N})_{n \in \mathbb{N}}$ converges to $f + \mathcal{N}$ in the norm $\|\cdot\|_2$. To this end let $\varepsilon > 0$ and choose $N \in \mathbb{N}$ such that $\|f_n - f_m\|_2 < \varepsilon$ for $n, m \ge N$. By Fatou's lemma one obtains

$$\int_{\mathbb{R}^d} |f_n - f|^2 d\lambda \leqslant \liminf_{m \to \infty} \int_{\mathbb{R}^d} |f_n - f_m|^2 d\lambda \leqslant \varepsilon^2 \quad \text{for all } n \ge N \;.$$

Hence $\lim_{n\to\infty} ||f_n - f||_2 = 0$, and $L^2(\mathbb{R}^d)$ endowed with the inner product $\langle \cdot, \cdot \rangle$ is a Hilbert space. It is called the *Hilbert space of square-integrable functions* on \mathbb{R}^d . Note that for every complete measure space (Ω, μ) one obtains in the same way the Hilbert space $L^2(\Omega, \mu)$ of square-integrable functions on (Ω, μ) .

3.1.10 Theorem Let V be a normed \mathbb{K} -vector space. Then the norm $\|\cdot\| : V \to \mathbb{R}_{\geq 0}$ is associated to an inner product $\langle \cdot, \cdot \rangle : V \times V \to \mathbb{K}$ if and only if the parallelogram identity

$$||v + w||^2 + ||v - w||^2 = 2||v||^2 + 2||w||^2$$

holds true for all $v, w \in V$. In this case, the inner product of two elements $v, w \in V$ can be expressed by the polarization identity for $\mathbb{K} = \mathbb{R}$

$$\langle v, w \rangle = \frac{1}{4} \left(\|v + w\|^2 - \|v - w\|^2 \right) = \frac{1}{2} \left(\|v + w\|^2 - \|v\|^2 - \|w\|^2 \right)$$
 (A.3.1.9)

respectively by the polarization identity for $\mathbb{K} = \mathbb{C}$

$$\langle v, w \rangle = \frac{1}{4} \sum_{k=1}^{4} i^{k} ||w + i^{k} v||^{2}$$
 (A.3.1.10)

Proof. The forward direction is a consequence of 3.1.4, Eq. A.3.1.3. To show the backward direction we consider two cases $\mathbb{K} = \mathbb{R}$ and $\mathbb{K} = \mathbb{C}$ separately.

1. Case. Given the norm $\|\cdot\|$ define $\langle\cdot,\cdot\rangle: V \times V \to \mathbb{R}$ by real polarization

$$\langle v, w \rangle = \frac{1}{4} \left(\|v + w\|^2 - \|v - w\|^2 \right), \text{ where } v, w \in \mathcal{V} \ .$$

Note that the parallelogram identity entails

$$\frac{1}{4} \left(\|v+w\|^2 - \|v-w\|^2 \right) = \frac{1}{2} \left(\|v+w\|^2 - \|v\|^2 - \|w\|^2 \right) .$$

Observe that by definition $\langle v, w \rangle = \langle w, v \rangle$ and $||v|| = \sqrt{\langle v, v \rangle}$. Let us show additivity in the first variable. Let $v_1, v_2, w \in V$ and compute using the parallelogram identity

$$\|v_1 + v_2 + w\|^2 = 2\|v_1 + w\|^2 + 2\|v_2\|^2 - \|v_1 + w - v_2\|^2 ,$$

$$\|v_1 + v_2 + w\|^2 = 2\|v_2 + w\|^2 + 2\|v_1\|^2 - \|v_2 + w - v_1\|^2 .$$

Hence

$$\|v_1 + v_2 \pm w\|^2 = \|v_1 \pm w\|^2 + \|v_2 \pm w\|^2 + \|v_1\|^2 + \|v_2\|^2 - \|v_1 \pm w - v_2\|^2 - \|v_2 \pm w - v_1\|^2.$$

Subtracting the - version from the + version of this equation entails

$$\langle v_1 + v_2, w \rangle = \frac{1}{4} \left(\|v_1 + v_2 + w\|^2 - \|v_1 + v_2 - w\|^2 \right) =$$

= $\frac{1}{4} \left(\|v_1 + w\|^2 + \|v_2 + w\|^2 - \|v_1 - w\|^2 - \|v_2 - w\|^2 \right) = \langle v_1, w \rangle + \langle v_2, w \rangle ,$

so additivity in the first variable is proved. By induction one derives from this that for all natural n

$$\langle nv, w \rangle = n \langle v, w \rangle$$
 for all $v, w \in V$. (A.3.1.11)

Since then $\langle -nv, w \rangle - n \langle v, w \rangle = \langle -nv + nv, w \rangle = 0$ for all $n \in \mathbb{N}$, Eq. (A.3.1.11) also holds for $n \in \mathbb{Z}$. Now let $p \in \mathbb{Z}$ and $q \in \mathbb{N}_{>0}$. Then $q \langle \frac{p}{q}v, w \rangle = \langle pv, w \rangle = p \langle v, w \rangle$, hence one has for rational r

$$\langle rv, w \rangle = r \langle v, w \rangle$$
 for all $v, w \in V$. (A.3.1.12)

Since addition, multiplication by scalars and the norm are continuous, the function

$$\mathbb{R} \to \mathbb{R}, \ r \mapsto \langle rv, w \rangle - r \langle v, w \rangle = \frac{1}{4} \left(\|rv + w\|^2 + r\|v - w\|^2 - \|rv - w\|^2 - r\|v + w\|^2 \right)$$

is continuous. Since it vanishes over \mathbb{Q} , it has to coincide with the zero map. Therefore, Eq. (A.3.1.12) holds for all $r \in \mathbb{R}$. So $\langle \cdot, \cdot \rangle$ is linear in the first coordinate. By symmetry, it is so too in the second coordinate. Hence $\langle \cdot, \cdot \rangle$ is a symmetric bilinear form inducing $\|\cdot\|$.

2. Case. In the case $\mathbb{K} = \mathbb{C}$ use complex polarization and put

$$\left\langle v,w\right\rangle = \frac{1}{4}\sum_{k=1}^4 \,\mathrm{i}^k\,\|w+\mathrm{i}^k\,v\|^2\quad\text{for all }v,w\in\mathrm{V}\ .$$

Then $\langle \cdot, \cdot \rangle$ is conjugate-symmetric, since

$$\overline{\langle v, w \rangle} = \frac{1}{4} \sum_{k=1}^{4} (-\mathbf{i})^k \|w + \mathbf{i}^k v\|^2 = \frac{1}{4} \sum_{k=1}^{4} (-\mathbf{i})^k \|(-\mathbf{i})^k w + v\|^2 = \langle w, v \rangle.$$

Next compute

$$\mathfrak{Re}\langle v,w\rangle = \frac{1}{4}\left(\|w+v\|^2 - \|w-v\|^2\right)$$

and

$$\Im \mathfrak{m} \langle v, w \rangle = \frac{1}{4} \left(\|w + \mathrm{i} v\|^2 - \|w - \mathrm{i} v\|^2 \right) \ .$$

By the first case one concludes that $\mathfrak{Re}\langle\cdot,\cdot\rangle$ and $\mathfrak{Im}\langle\cdot,\cdot\rangle$ are both \mathbb{R} -linear in the first and the second coordinate. Moreover,

$$\mathfrak{Re}\langle v,\mathsf{i}w\rangle = \frac{1}{4}\left(\|\mathsf{i}w + v\|^2 - \|\mathsf{i}w - v\|^2\right) = \frac{1}{4}\left(\|w - \mathsf{i}v\|^2 - \|w + \mathsf{i}v\|^2\right) = -\mathfrak{Im}\langle v,w\rangle = \mathfrak{Re}\,\mathsf{i}\langle v,w\rangle$$

and

$$\mathfrak{Im}\langle v, \mathsf{i}w\rangle = \frac{1}{4} \left(\|\mathsf{i}w + \mathsf{i}v\|^2 - \|\mathsf{i}w - \mathsf{i}v\|^2 \right) = \mathfrak{Re}\langle v, w\rangle = \mathfrak{Im} \,\mathsf{i}\langle v, w\rangle \,,$$

hence $\langle \cdot, \cdot \rangle$ is complex linear in the second coordinate. Finally,

$$\mathfrak{Re}\langle v,v\rangle = \|v\|^2 \quad \text{and} \quad \mathfrak{Im}\langle v,v\rangle = \frac{1}{4}\left(\|v+\mathsf{i} v\|^2 - \|v-\mathsf{i} v\|^2\right) = 0 \ .$$

This finishes the proof that $\langle \cdot, \cdot \rangle$ is a complex inner product inducing the norm $\|\cdot\|$.

3.1.11 Next we will turn Hilbert spaces into a category. To this end one needs to know what morphisms in this category should be. There are two options each giving rise to a category of Hilbert spaces. These categories just differ by their morphism classes. The first one is to have as morphisms linear maps $A : \mathcal{H}_1 \to \mathcal{H}_2$ preserving the inner products which means that they fulfill

$$\langle Av_1, Av_2 \rangle = \langle v_1, v_2 \rangle$$
 for all $v_1, v_2 \in \mathcal{H}_1$.

By Theorem 3.1.10 this property is equivalent to

$$||Av|| = ||v||$$
 for all $v \in \mathcal{H}_1$,

that is to A being norm preserving or isometric. Obviously, the identity map on a Hilbert space is isometric and the composition of two composable isometric linear maps is again isometric and linear. Hence Hilbert spaces together with norm preserving linear maps between them form a category which we denote by Hilb_{np}. The isomorphisms in this category are the surjective isometric linear maps between Hilbert spaces. Such maps are called *unitary*. The condition of a linear map being norm preserving is pretty restrictive, so the category Hilb_{np} contains only few morphisms. This can be cured by allowing all *bounded* linear maps between Hilbert spaces to be morphisms that is of all linear $A: \mathcal{H}_1 \to \mathcal{H}_2$ for which there exists a $C \ge 0$ such that

$$||Av|| \leq C ||v|| \quad \text{for all } v \in \mathcal{H}_1 . \tag{A.3.1.13}$$

The existence of a smallest such C is guaranteed by the following. It is called the *operator norm* of A and is denoted ||A||.

3.1.12 Lemma The operator norm of a bounded linear operator $A : \mathcal{H}_1 \to \mathcal{H}_2$ between Hilbert spaces \mathcal{H}_1 and \mathcal{H}_2 exists and is given by

$$||A|| = \sup \{ ||Av|| \mid v \in \mathcal{H}_1, ||v|| = 1 \}$$

= sup { ||Av|| \mid v \in \mathcal{H}_1, ||v|| \leq 1 }
= sup { |\langle w, Av \rangle | \mid v \in \mathcal{H}_1, w \in \mathcal{H}_2, ||v|| = ||w|| = 1 }.

Proof. If $A : \mathcal{H}_1 \to \mathcal{H}_2$ is bounded, then the set $\{ \|Av\| \mid v \in \mathcal{H}_1, \|v\| = 1 \}$ is bounded, hence has a supremum C_0 . This implies that for all non-zero $v \in \mathcal{H}_1$

$$||Av|| = ||v|| \left\| A\left(\frac{v}{||v||}\right) \right\| \le C_0 ||v||$$
.

Hence the estimate (A.3.1.13) holds true for $C = C_0$. Moreover, C_0 is the smallest such C because if $0 \leq C_1 < C_0$, then there exists $v \in \mathcal{H}_1$ with ||v|| = 1 and $||Av|| > C_1$. This proves that the operator norm of A exists and that it fulfills $||A|| = C_0$.

By definition of C_0 , the estimate $||A|| = C_0 \leq \sup \{||Av|| \mid v \in \mathcal{H}_1, ||v|| \leq 1\}$ holds true. By definition of the operator norm, $||Av|| \leq ||A||$ for all $v \in \mathcal{H}_1$ with $||v|| \leq 1$. The two estimates together entail the equality $||A|| = \sup \{||Av|| \mid v \in \mathcal{H}_1, ||v|| \leq 1\}$.

The Cauchy–Schwarz inequality entails

$$\sup\left\{\left|\langle w, Av\rangle\right| \mid v \in \mathcal{H}_1, \ w \in \mathcal{H}_2, \ \|v\| = \|w\| = 1\right\} \leqslant \|A\|$$

The converse estimate follows by the observation that

$$\sup\left\{\left|\langle w, Av\rangle\right| \mid w \in \mathcal{H}_{2}, \, \|w\| = 1\right\} \geqslant \left|\left\langle\frac{Av}{\|Av\|}, Av\right\rangle\right| = \|Av\|$$

whenever $Av \neq 0$. This proves the last claimed equality.

Every norm preserving linear map is bounded with operator norm 1. In particular, the identity map on a Hilbert space is bounded. Moreover, if $A : \mathcal{H}_1 \to \mathcal{H}_2$ and $B : \mathcal{H}_2 \to \mathcal{H}_3$ are bounded linear operators between Hilbert spaces, then the composition $BA : \mathcal{H}_1 \to \mathcal{H}_3$ is bounded with operator norm $\leq \|B\| \|A\|$ since for all $v \in \mathcal{H}_1$ with $\|v\| \leq 1$

$$||BAv|| \leq ||B|| ||Av|| \leq ||B|| ||A||$$
.

Hence Hilbert spaces as objects together with bounded linear maps as morphisms form a category which we denote by Hilb and call the *category of Hilbert spaces*. Note that the morphisms in this category appear to "forget" the inner product and just preserve the linear and the topological structure. John Baez (Baez, 1997, p. 133) has explained how to heal this apparent defect by showing that Hilb carries a so-called *-structure given by the adjoint map on bounded linear operators. We will come back to this point later when we introduce adjoint operators.

As proved already for Banach spaces, a linear map between Hilbert spaces is bounded if and only if it is continuous. For reasons of completeness and convenience we state here the result for inner product spaces.

3.1.13 Proposition Let $A : \mathcal{H}_1 \to \mathcal{H}_2$ be a linear map between two inner product spaces. Then the following are equivalent.

- (i) A is bounded.
- (ii) A is continuous.
- (iii) A is continuous at 0.

Proof. (i) \implies (ii). Assume that A is bounded. Let $||A|| := \sup_{v \in \mathcal{H}_1} ||Av||$ be its norm. Then, for all $v, w \in \mathcal{H}_1$

$$||Av - Aw|| \le ||A|| \cdot ||v - w||$$
.

Hence A is Lipschitz continuous, so in particular continuous. (ii) \implies (iii). If the map A is continuous, it is in particular continuous at the origin. (iii) \implies (i). If A is continuous at the origin, there exists $\delta > 0$ such that for all $v \in \mathcal{H}_1$ the estimate ||Av|| < 1 holds whenever $||v|| < \delta$. This implies that for v with $||v|| \leq 1$

$$\|Av\| = 2\delta \left\| A\left(\frac{1}{2\delta}v\right) \right\| < 2\delta$$

This means that A is bounded.

3.1.14 Last in this section we will introduce bounded bilinear and sesquilinear maps. We define them for normed vector spaces. Their main application lies in the operator theory on Hilbert spaces, so we introduce them here.

3.1.15 Definition Let V_1 and V_2 be two normed vector spaces over \mathbb{K} and denote the norms on V_1 and V_2 by the same symbol $\|\cdot\|$. Assume that $b: V_1 \times V_2 \to \mathbb{K}$ is a *bilinear* or *sesquilinear form* that is b is linear in each argument respectively b is conjugate linear in the first and linear in the second argument. The form $b: V_1 \times V_2 \to \mathbb{K}$ then is called *bounded* if there exists a C > 0 such that

$$|b(v,w)| \leq C ||v|| ||w||$$
 for all $v \in V_1, w \in V_2$.

In this case,

$$|b\| := \sup \{ |b(v, w)| \mid v \in \mathcal{V}_1, \ w \in \mathcal{V}_2, \ \|v\| = \|w\| = 1 \}$$

exists and is called the *norm* of the form b.

3.1.16 Example The inner product on a (pre-) Hilbert space is bounded by the Cauchy–Schwarz inequality and has norm 1.

3.1.17 Proposition A bilinear or sesquilinear form $b : V_1 \times V_2 \to \mathbb{K}$ defined on the cartesian product of two normed vector space V_1 and V_2 over \mathbb{K} is bounded if and only if it is continuous.

Proof. If b is bounded, then

$$\begin{aligned} |b(v,w) - b(v',w')| &\leq |b(v,w) - b(v',w)| + |b(v',w) - b(v',w')| \leq \\ &\leq \|b\| (\|w\| \|v - v'\| + \|v'\| \|w - w'\|) \end{aligned}$$

for all $v, v' \in V_1$ and $w, w' \in V_2$. Hence b is locally Lipschitz continuous, so in particular continuous. Conversely, assume now that b is continuous. Then one can find $\delta > 0$ such that for all $v \in V_1$ and $w \in V_2$ of norm less than δ the relation |b(v, w)| < 1 holds true. But that entails for all non-zero v, w

$$b(v,w)| = \frac{4 \|v\| \|w\|}{\delta^2} \cdot b\left(\delta \frac{v}{2\|v\|}, \delta \frac{w}{2\|w\|}\right) \leq \frac{4}{\delta^2} \|v\| \|w\|.$$

Hence b is bounded.

3.1.18 Remark Given two normed vector spaces or more generally two topological vector spaces V_1 and V_2 one can consider bilinear or sesquilinear forms $b: V_1 \times V_2 \to \mathbb{K}$ which are only *separatelycontinuous*. That means that for all $v \in V_1$ the map $b_v = b(v, -): V_2 \to \mathbb{K}$ and for all $w \in V_2$ the map $\overline{b}_w = b(-, w): V_1 \to \mathbb{K}$ is continuous. In general, separate-continuity is strictly weaker than continuity unless the underlying vector spaces are Banach spaces where the two notions coincide as a consequence of the Banach–Steinhaus theorem. Let us prove this. By continuity of b_v there exist $C_v \ge 0$ such that $|b_v(w)| \le C_v ||w||$ for all $w \in V_2$ and $\overline{C}_w \ge 0$ such that $|\overline{b}_w(v)| \le \overline{C}_w ||v||$ for all $v \in V_1$. Hence, for all $w \in V_2$

$$\sup_{v \in \mathcal{V}, \|v\| \leq 1} |b_v(w)| = \sup_{v \in \mathcal{V}, \|v\| \leq 1} |\overline{b}_w(v)| \leq \overline{C}_w < \infty .$$

The Banach-Steinhaus theorem now entails

$$\sup_{v,w\in \mathcal{V}, \, \|v\|, \, \|w\| \leq 1} |b(v,w)| = \sup_{v\in \mathcal{V}, \, \|v\| \leq 1} \|b_v\| < \infty \; .$$

Therefore, *b* is bounded, so continuous by the preceding proposition.

A.3.2. Orthogonal decomposition and the Riesz representation theorem

3.2.1 One of the issues with infinite-dimensional analysis is that a closed subspace of an infinite dimensional Banach space might not have a closed complement. Fortunately, the situation in Hilbert space theory is not so grim because every closed subspace of a Hilbert space admits an orthogonal complement. This is one of the four crucial properties which distinguish Hilbert spaces from Banach spaces and which are stated in the following.

In this section \mathcal{H} will always denote a Hilbert space over the field $\mathbb{K} = \mathbb{R}$ or $\mathbb{K} = \mathbb{C}$. The symbol $\langle \cdot, \cdot \rangle$ will stand for the inner product of \mathcal{H} .

3.2.2 Theorem (Best approximation theorem) Every closed convex nonempty subset C of a Hilbert space \mathcal{H} has a unique element of minimal norm.

Proof. Let $d = \inf\{||v|| \mid v \in C\}$ which is a non-negative real number. We claim there exists a unique $v_0 \in C$ with $||v_0|| = d$. For uniqueness, consider two vectors v_0, v_1 satisfying the desired property, and let $v = \frac{1}{2}(v_0 + v_1)$ be their midpoint. Then

$$\|v\| = \frac{1}{2} \|v_0 + v_1\| \leq \frac{1}{2} (\|v_0\| + \|v_1\|) = d$$

By minimality of d this entails ||v|| = d. By the parallelogram identity

$$\left\|\frac{1}{2}(v_0+v_1)\right\|^2 + \left\|\frac{1}{2}(v_0-v_1)\right\|^2 = 2\left\|\frac{v_0}{2}\right\|^2 + 2\left\|\frac{v_1}{2}\right\|^2 = d^2,$$

hence

$$\left\|\frac{1}{2}(v_0 - v_1)\right\|^2 \leqslant d^2 - \|v\|^2 = 0 ,$$

proving $v_0 = v_1$.

For the proof of existence observe that by definition of d there exists a sequence $(v_n)_{n \in \mathbb{N}} \subset C$ such that $\lim_{n \to \infty} ||v_n|| = d$. By convexity

$$\frac{1}{2}(v_n + v_m) \in C$$

for all $n,m\in\mathbb{N}$, hence $\frac{1}{4}\|v_n+v_m\|^2\geqslant d^2$. The parallelogram equality entails

$$0 \leq ||v_n - v_m||^2 = 2||v_n||^2 + 2||v_m||^2 - ||v_n + v_m||^2 \leq 2||v_n||^2 + 2||v_m||^2 - 4d^2.$$

Since $\lim_{n\to\infty} \|v_n\| = d$ there exists for given $\varepsilon > 0$ an $N \in \mathbb{N}$ such that $\|v_n\|^2 - d^2 \leq \frac{1}{4}\varepsilon^2$ for all $n \ge N$. Hence, for $n, m \ge N$

$$0 \leqslant \|v_n - v_m\| \leqslant \varepsilon ,$$

and $(v_n)_{n \in \mathbb{N}}$ is a Cauchy-sequence, so convergent by completeness of \mathcal{H} . Put $v_0 := \lim_{n \to \infty} v_n$. Then $v_0 \in C$ since C is closed and $||v_0|| = \lim_{n \to \infty} ||v_n|| = d$. The existence claim follows and the proof is finished.

3.2.3 Theorem and Definition (Orthogonal decomposition theorem) Let $V \subset \mathcal{H}$ be a closed subspace of the Hilbert space \mathcal{H} . Then the orthogonal complement

$$\mathbf{V}^{\perp} = \left\{ w \in \mathcal{H} \mid \langle v, w \rangle = 0 \text{ for each } v \in \mathbf{V} \right\}$$

is a closed subspace of \mathfrak{H} and $\mathfrak{H} = V \oplus V^{\perp}$. The map $\operatorname{pr}_{V} : \mathfrak{H} \to V$ which maps $w \in \mathfrak{H}$ to the unique $w_1 \in V$ such that $w - w_1 \in V^{\perp}$ is called the orthogonal projection onto V. It satisfies $||w - \operatorname{pr}_{V}(w)|| = d(w, V) := \inf \{||v - w|| \mid v \in V\}$ that is $\operatorname{pr}_{V}(w)$ is the unique element of V having shortest distance from w.

Proof. For $v \in \mathcal{H}$ define $v^{\flat} : \mathcal{H} \to \mathbb{R}$ by $v^{\flat}(w) = \langle w, v \rangle$. Recall that this map is continuous and linear. Hence the kernel $(v^{\flat})^{-1}(0)$ is a closed linear subspace of \mathcal{H} and

$$\mathbf{V}^{\perp} = \bigcap_{v \in \mathbf{V}} (v^{\flat})^{-1}(0)$$
 (A.3.2.1)

is a closed linear subspace. To show $V \cap V^{\perp} = \{0\}$, consider $v \in V \cap V^{\perp}$. Then $||v||^2 = \langle v, v \rangle = 0$. Next we want to show that every $w \in \mathcal{H}$ can be written in the form $w = w_1 + w_2$ with $w_1 \in V$ and $w_2 \in V^{\perp}$. To see this put C = w - V. Then C is closed and convex. By the best approximation theorem there exists a unique element $w_2 \in C$ of minimal norm. Let w_1 be the unique element of V such that $w_2 = w - w_1$. It remains to show $w_2 \in V^{\perp}$. Since w_2 has minimal norm among the elements of w - V, the following inequality holds for all vectors $v \in V$:

$$||w_2||^2 \leq ||w_2 + v||^2 = ||w_2||^2 + 2 \Re(w_2, v) + ||v||^2$$

Hence

$$0 \leq 2 \Re(w_2, v) + ||v||^2$$
 for all $v \in V$.

Now assume that ||v|| = 1 and choose $\varphi \in \mathbb{R}$ such that $e^{i\varphi} \langle w_2, v \rangle \in \mathbb{R}$. Setting $v' = e^{i\varphi}v$, one obtains for all $\lambda \in \mathbb{R}$ by the last inequality

$$0 \leq 2\langle w_2, \lambda v' \rangle + \|\lambda v'\|^2 = 2\lambda \langle w_2, v' \rangle + \lambda^2 .$$

For $\lambda=-\langle w_2,v'
angle$ this entails the estimate

$$|\langle w_2, v' \rangle|^2 = -\left(-2|\langle w_2, v' \rangle|^2 + |\langle w_2, v' \rangle|^2\right) = -\left(2\lambda \langle w_2, v' \rangle + \lambda^2\right) \le 0$$

Hence $\langle w_2, v \rangle = 0$ for all unit vectors $v \in V$, therefore $w_2 \in V^{\perp}$.

The remainder of the claim is now an immediate consequence of the construction of w_1 from the given w and the observation that $pr_V(w) = w_1$.

3.2.4 Corollary For every subspace $V \subset \mathcal{H}$ of a Hilbert space \mathcal{H} the orthogonal complement V^{\perp} is closed, and the relation $V^{\perp} - \overline{V}^{\perp}$

holds true. Moreover,

$$\overline{V} = (V^{\perp})^{\perp}$$

Proof. By Equation (A.3.2.1), the orthogonal complement V^{\perp} is closed. Since $V \subset \overline{V}$ the inclusion $\overline{V}^{\perp} \subset V^{\perp}$ holds true. The converse inclusion $V^{\perp} \subset \overline{V}^{\perp}$ follows from the observation that if $w \in V^{\perp}$ and $(v_n)_{n \in \mathbb{N}}$ is a sequence in V converging to some $v \in \overline{V}$, then

$$\langle w, v \rangle = \lim_{n \to \infty} \langle w, v_n \rangle = 0$$
.

This proves the equality $V^{\perp} = \overline{V}^{\perp}$. The inclusion $\overline{V} \subset (\overline{V}^{\perp})^{\perp} = (V^{\perp})^{\perp}$ is immediate by definition of the orthogonal complement. Since

$$\mathcal{H} = \overline{V} \oplus V^{\perp} = (V^{\perp})^{\perp} \oplus V^{\perp}$$

by the preceding theorem, the equality $\overline{V}=(V^{\perp})^{\perp}$ follows.

3.2.5 Theorem (Riesz representation theorem for Hilbert spaces) Let \mathcal{H} be a Hilbert space and \mathcal{H}' its topological dual. Then the musical map

$${}^{\flat}: \mathcal{H} \to \mathcal{H}', \quad v \mapsto v^{\flat} = (\mathcal{H} \ni w \mapsto \langle v, w \rangle \in \mathbb{K})$$

is an isometric isomorphism which is linear in the real case and conjugate-linear in the complex case.

Proof. Obviously, \flat is linear if the ground field \mathbb{K} equals \mathbb{R} and conjugate-linear if $\mathbb{K} = \mathbb{C}$. Now observe that for all $v \in \mathcal{H}$ by the Cauchy–Schwarz inequality

$$\|v^{\flat}\| = \sup\left\{|\langle v, w \rangle| \mid w \in \mathcal{H} \& \|w\| = 1\right\} = \|v\|,$$

hence ${}^{\flat}$ is an isometry, so in particular injective. It remains to show surjectivity. So assume that $\alpha : \mathcal{H} \to \mathbb{K}$ is a nontrivial continuous linear form. Let V be its kernel. Then V is a closed linear subspace of \mathcal{H} . Since α is nontrivial, the orthogonal complement V^{\perp} is nontrivial, too. Hence $V^{\perp} \cong \mathcal{H}/V$ is isomorphic to $\operatorname{im} \alpha = \mathbb{K}$ and there exists a vector $v \in V^{\perp} \setminus \{0\}$ such that $\alpha(v) = 1$. Since v spans V^{\perp} there exists for every $w \in \mathcal{H}$ a unique $\lambda_w \in \mathbb{K}$ such that $w = \operatorname{pr}_V(w) + \lambda_w v$. Then compute

$$\alpha(w) = \alpha(\lambda_w v) = \lambda_w \quad \text{and} \quad \left(\frac{v}{\|v\|^2}\right)^\flat(w) = \frac{1}{\|v\|^2} \langle v, w \rangle = \frac{\lambda_w}{\|v\|^2} \langle v, v \rangle = \lambda_w \ .$$

This entails $\alpha = \left(\frac{v}{\|v\|^2}\right)^{\flat}$, and $^{\flat}$ is surjective.

3.2.6 Remark Sometimes in the Hilbert space literature the inverse of the musical isomorphism ${}^{\flat}: \mathcal{H} \to \mathcal{H}'$ is denoted ${}^{\sharp}: \mathcal{H}' \to \mathcal{H}$. We will follow that convention.

3.2.7 Corollary Every Hilbert space \mathcal{H} is reflexive that is the canonical map

$$H \to H'', v \mapsto (H' \ni \lambda \mapsto \lambda(v) \in \mathbb{K})$$

is an isometric isomorphism.

Proof. By the Riesz Representation Theorem, the dual \mathcal{H}' is a Hilbert space with inner product

$$\langle\!\langle\!\langle\cdot,\cdot\rangle\!\rangle: \mathcal{H}' \times \mathcal{H}' \to \mathbb{K}, \ (\lambda,\mu) \mapsto \langle\!\langle\lambda,\mu\rangle\!\rangle = \langle\mu^{\sharp},\lambda^{\sharp}\rangle.$$

Hence, by applying the Riesz Representation Theorem twice, the map ${}^{\flat} \circ {}^{\flat} : \mathcal{H} \to \mathcal{H}''$ is an isometric linear isomorphism. Now compute for $v \in \mathcal{H}$ and $\lambda \in \mathcal{H}'$

$$(v^{\flat})^{\flat}(\lambda) = \langle\!\langle v^{\flat}, \lambda \rangle\!\rangle = \langle \lambda^{\sharp}, v \rangle = \lambda(v) \;.$$

Hence ${}^{\flat} \circ {}^{\flat}$ coincides with the canonical map above and the claim follows.

3.2.8 Corollary Let \mathcal{H}_1 and \mathcal{H}_2 be two Hilbert spaces and $b : \mathcal{H}_1 \times \mathcal{H}_2 \to \mathbb{K}$ a bounded sesquilinear form. Then there exists unique bounded linear map $A : \mathcal{H}_2 \to \mathcal{H}_1$ such that

$$b(v,w) = \langle v, Aw \rangle$$
 for all $v \in \mathcal{H}_1, w \in \mathcal{H}_2$. (A.3.2.2)

Moreover, the operator norm ||A|| coincides with ||b||.

Proof. First let us show uniqueness. So let $A, B : \mathcal{H}_2 \to \mathcal{H}_1$ be bounded and linear so that

$$b(v,w) = \langle v, Aw \rangle = \langle v, Bw \rangle$$
 for all $v \in \mathcal{H}_1, w \in \mathcal{H}_2$.

Then $||(A-B)w||^2 = \langle (A-B)w, Aw - Bw \rangle = b((A-B)w, w) - b((A-B)w, w) = 0$ for all $w \in \mathcal{H}_2$ which entails equality of A and B.

To prove existence observe that for every $w \in \mathcal{H}_2$ the map

$$\overline{b}_w : \mathcal{H}_1 \to \mathbb{K}, \ v \mapsto \overline{b}(w, v) := b(v, w)$$

is bounded and linear, so by the Riesz representation theorem there exists for every w a unique element $Aw \in \mathcal{H}_1$ such that $\langle Aw, v \rangle = \overline{b}(w, v)$ for all $v \in \mathcal{H}_1$. By construction, $Aw = (\overline{b}_w)^{\sharp}$. Since the maps $\mathcal{H}_2 \to \mathcal{H}'_1$, $w \mapsto \overline{b}_w$ and $^{\sharp} : \mathcal{H}'_1 \to \mathcal{H}_1$ are both conjugate-linear, A is linear. Hence A is the desired linear operator fulfilling Equation (A.3.2.2).

For the operator norm compute

$$||A|| = \sup \{ |\langle v, Aw \rangle| \mid v \in \mathcal{H}_1, \ w \in \mathcal{H}_2, \ ||v|| = ||w|| = 1 \} = = \sup \{ |b(v, w)| \mid v \in \mathcal{H}_1, \ w \in \mathcal{H}_2, \ ||v|| = ||w|| = 1 \} = ||b||.$$

Hence A is bounded with operator norm equal to ||b|| and the claim is proved.

3.2.9 Last in this section we will examine the *Hilbert direct sum* or just *Hilbert sum* of a family $(\mathcal{H}_i)_{i \in I}$ of Hilbert spaces. It is defined by

$$\begin{split} \widehat{\bigoplus}_{i \in I} \mathcal{H}_i &= \left\{ (v_i)_{i \in I} \in \prod_{i \in I} \mathcal{H}_i \mid \left(\|v_i\|^2 \right)_{i \in I} \text{ is summable} \right\} = \\ &= \left\{ (v_i)_{i \in I} \in \prod_{i \in I} \mathcal{H}_i \mid \exists C \ge 0 \,\forall J \in \mathcal{P}_{\mathsf{fin}}(I) : \sum_{i \in J} \|v_i\|^2 \leqslant C \right\} \end{split}$$

where, as usual, $\mathcal{P}_{fin}(I)$ denotes the set of all finite subsets of I.

3.2.10 Proposition Let $(\mathcal{H}_i)_{i \in I}$ be a family of Hilbert spaces. Then the Hilbert direct sum $\bigoplus_{i \in I} \mathcal{H}_i$ is a Hilbert space with inner product given by

$$\langle -, - \rangle : \bigoplus_{i \in I} \mathcal{H}_i \times \bigoplus_{i \in I} \mathcal{H}_i \to \mathbb{K}, \quad ((v_i)_{i \in I}, (w_i)_{i \in I}) \mapsto \sum_{i \in I} \langle v_i, w_i \rangle.$$

Proof. We show first that $\widehat{\bigoplus}_{i \in I} \mathcal{H}_i$ is a subvector space of the direct product $\prod_{i \in I} \mathcal{H}_i$. Let $z \in \mathbb{K}$ and $(v_i)_{i \in I}, (w_i)_{i \in I} \in \widehat{\bigoplus}_{i \in I} \mathcal{H}_i$. Choose $C, D \ge 0$ such that

$$\sum_{i \in J} \|v_i\|^2 \leqslant C \quad \text{and} \quad \sum_{i \in J} \|w_i\|^2 \leqslant D \quad \text{for all } J \in \mathcal{P}_{\mathsf{fin}}(I) \ .$$

Then

$$\sum_{i \in J} \|zv_i\|^2 = |z| \sum_{i \in J} \|v_i\|^2 \le |z| C \quad \text{for all } J \in \mathcal{P}_{\text{fin}}(I) , \qquad (A.3.2.3)$$

so $(zv_i)_{i\in I}\in \bigoplus_{i\in I}\mathcal{H}_i$. Moreover, by Minkowski's inequality for finite sums,

$$\sum_{i \in J} \|v_i + w_i\|^2 \le \left(\sqrt{\sum_{i \in J} \|v_i\|^2} + \sqrt{\sum_{i \in J} \|w_i\|^2} \right)^2 \le \left(\sqrt{C} + \sqrt{D} \right)^2 \quad \text{for all } J \in \mathcal{P}_{\mathsf{fin}}(I) \ . \tag{A.3.2.4}$$

Hence the family $(||v_i + w_i||^2)_{i \in I}$ is summable and $(v_i + w_i)_{i \in I} \in \bigoplus_{i \in I} \mathcal{H}_i$.

Next observe that the map

$$\|-\|: \bigoplus_{i\in I} \mathcal{H}_i \to \mathbb{K}, \ (v_i)_{i\in I} \mapsto \|(v_i)_{i\in I}\| = \sqrt{\sum_{i\in I} \|v_i\|^2}$$

is well-defined by definition of the Hilbert direct sum. It is even a norm by (A.3.2.3) and (A.3.2.4). Now we need to show that the inner product on $\bigoplus_{i \in I} \mathcal{H}_i$ is well-defined which means that the family $(\langle v_i, w_i \rangle)_{i \in I}$ is summable for all $(v_i)_{i \in I}, (w_i)_{i \in I} \in \bigoplus_{i \in I} \mathcal{H}_i$. To this end let $J \subset I$ be a finite subset. Then, by the triangle inequality, the Cauchy–Schwarz inequality on the Hilbert spaces \mathcal{H}_i and the Cauchy–Schwarz inequality for finite sums,

$$\left|\sum_{i\in J} \langle v_i, w_i \rangle\right| \leq \sum_{i\in J} \left| \langle v_i, w_i \rangle \right| \leq \sum_{i\in J} \|v_i\| \|w_i\| \leq \sqrt{\sum_{i\in J} \|v_i\|^2} \cdot \sqrt{\sum_{i\in J} \|w_i\|^2} \leq \left\| (v_i)_{i\in I} \right\| \|(w_i)_{i\in I}\|$$

Hence the family $(\langle v_i, w_i \rangle)_{i \in I}$ is absolutely summable, so in particular summable, and the inner product is well-defined.

By definition and since all the inner products on the Hilbert spaces \mathcal{H}_i are conjugate symmetric and positive definite, the map $\langle -, - \rangle$ on $\bigoplus_{i \in I} \mathcal{H}_i$ has to be conjugate symmetric and positive definite as well. It remains to show linearity in the second argument. Denote for $(v_i)_{i \in I}, (w_i)_{i \in I}, (w_i)_{i \in I} \in \prod_{i \in I} \mathcal{H}_i$ and $J \in \mathcal{P}_{\text{fin}}(I)$ by $\langle (v_i)_{i \in I}, (w_i)_{i \in I} \rangle_J$ the finite sum $\sum_{i \in J} \langle v_i, w_i \rangle$. Observe that the net $(\langle (v_i)_{i \in I}, (w_i)_{i \in I}, (w_i)_{i \in I}, (w_i)_{i \in I}, (w_i)_{i \in I} \rangle)_{J \in \mathcal{P}_{\text{fin}}(I)}$ converges to $\langle (v_i)_{i \in I}, (w_i)_{i \in I} \rangle$ in case both $(v_i)_{i \in I}$ and $(w_i)_{i \in I}$ are in $\bigoplus_{i \in I} \mathcal{H}_i$. Now let $z \in \mathbb{K}$ and $(v_i)_{i \in I}, (w_i)_{i \in I}, (w'_i)_{i \in I} \in \bigoplus_{i \in I} \mathcal{H}_i$. Then

$$\begin{array}{l} \langle (v_i)_{i \in I}, (w_i)_{i \in I} + (w'_i)_{i \in I} \rangle_J = \langle (v_i)_{i \in I}, (w_i)_{i \in I} \rangle_J + \langle (v_i)_{i \in I}, (w'_i)_{i \in I} \rangle_J & \text{and} \\ \langle (v_i)_{i \in I}, z(w_i)_{i \in I} \rangle_J = z \langle (v_i)_{i \in I}, (w_i)_{i \in I} \rangle_J & . \end{array}$$

By convergence of all the nets $(\langle (v_i)_{i\in I}, (w_i)_{i\in I} \rangle_J)_{J \in \mathcal{P}_{fin}(I)}$, linearity in the second argument follows. By construction, the norm associated to the inner product $\langle -, - \rangle$ on $\bigoplus_{i\in I} \mathcal{H}_i$ coincides with the above defined norm ||-||. It remains to show that $\bigoplus_{i\in I} \mathcal{H}_i$ equipped with the norm ||-|| is complete. To this end observe that for every finite $J \subset I$ the map

$$\|-\|_J:\prod_{i\in I}\mathcal{H}_i\to\mathbb{R}_{\ge 0},\ (v_i)_{i\in I}\mapsto\sqrt{\langle (v_i)_{i\in I}, (v_i)_{i\in I}\rangle_J}=\sqrt{\sum_{i\in J}\|v_i\|^2}$$

is a seminorm and that $(v_i)_{i \in I} \in \prod_{i \in I} \mathcal{H}_i$ lies in the Hilbert direct sum $\widehat{\bigoplus}_{i \in I} \mathcal{H}_i$ if and only if the family $(\|(v_i)_{i \in I}\|_J)_{J \in \mathcal{P}_{\text{fin}}(I)}$ is bounded. Now let $((v_i^n)_{i \in I})_{n \in \mathbb{N}}$ be a Cauchy sequence. Let $\varepsilon > 0$ and choose $N_{\varepsilon} \in \mathbb{N}$ such that

$$\|(v_i^m)_{i\in I} - (v_i^n)_{i\in I}\| < \varepsilon \quad \text{for all } n, m \ge N_{\varepsilon} .$$
(A.3.2.5)

Then

$$\left\| (v_i^m)_{i \in I} - (v_i^n)_{i \in I} \right\|_J < \varepsilon \quad \text{for all } J \in \mathcal{P}_{\mathsf{fin}}(I) \text{ and } n, m \ge N_{\varepsilon} . \tag{A.3.2.6}$$

Taking $J = \{j\}$ for $j \in I$ this implies that the sequence $(v_j^n)_{n \in \mathbb{N}}$ is a Cauchy sequence in the Hilbert space \mathcal{H}_j . Let $v_j \in \mathcal{H}_j$ be its limit. The family $(v_i)_{i \in I}$ then is an element of $\bigoplus_{i \in I} \mathcal{H}_i$. To verify this put $N = N_1$ and observe that by (A.3.2.6) for all finite $J \subset I$

$$\| (v_i)_{i \in I} \|_J \leq \| (v_i^N)_{i \in I} \|_J + \| (v_i)_{i \in I} - (v_i^N)_{i \in I} \|_J = = \| (v_i^N)_{i \in I} \|_J + \lim_{m \to \infty} \| (v_i^m)_{i \in I} - (v_i^N)_{i \in I} \|_J \leq \| (v_i^N)_{i \in I} \| + 1.$$

Hence the family $(\|(v_i)_{i \in I}\|_J)_{J \in \mathcal{P}_{fin}(I)}$ is bounded and $(v_i)_{i \in I}$ lies in the Hilbert direct sum of the spaces \mathcal{H}_i , $i \in I$. Moreover, (A.3.2.6) entails that

$$\left\| (v_i)_{i \in I} - (v_i^n)_{i \in I} \right\|_J = \lim_{m \to \infty} \left\| (v_i^m)_{i \in I} - (v_i^n)_{i \in I} \right\|_J \leqslant \varepsilon \quad \text{for all } J \in \mathcal{P}_{\mathsf{fin}}(I) \text{ and } n \geqslant N_{\varepsilon} \ .$$

Since $||(v_i)_{i \in I} - (v_i^n)_{i \in I}||$ is the limit of the net $(||(v_i)_{i \in I} - (v_i^n)_{i \in I}||_J)_{J \in \mathcal{P}_{fin}(I)}$, the estimate

$$|(v_i)_{i\in I} - (v_i^n)_{i\in I}|| \leq \varepsilon$$
 for all $n \ge N_{\varepsilon}$

follows, and the sequence $((v_i^n)_{i \in I})_{n \in \mathbb{N}}$ convergences to $(v_i)_{i \in I}$. This finishes the proof.

A.3.3. Orthonormal bases in Hilbert spaces

3.3.1 Definition A (possibly empty) subset S of a Hilbert space \mathcal{H} is called an *orthogonal system* or just *orthogonal* if for any two different elements $v, w \in S$ the relation $\langle v, w \rangle = 0$ holds true. If in addition ||v|| = 1 for all elements $v \in S$, then the set is called *orthonormal* or an *orthonormal system*. A family $(v_i)_{i \in I}$ of vectors in \mathcal{H} is called *orthogonal* if $\langle v_i, v_j \rangle = 0$ for all $i, j \in I$ with $i \neq j$ and *orthonormal* if in addition $||v_i|| = 1$ for all $i \in I$.

3.3.2 Obviously, the set of orthonormal subsets of a Hilbert space is ordered by set-theoretic inclusion. Therefore, the following definition makes sense.

3.3.3 Definition A maximal orthonormal set in a Hilbert space \mathcal{H} is called an *orthonormal basis* or a *Hilbert basis* of \mathcal{H} .

3.3.4 Proposition Every Hilbert space \mathcal{H} has an orthonormal basis.

Proof. Wothout loss of generality we can assume that $\mathcal{H} \neq \{0\}$, because \emptyset is a Hilbert basis for $\{0\}$. Let \mathcal{O} denote the set of orthonormal subsets of \mathcal{H} . As mentioned before, \mathcal{O} is ordered by set-theoretic inclusion. Let $\mathcal{C} \subset \mathcal{O}$ be a non-empty chain. Put $U = \bigcup_{S \in \mathcal{C}} S$. Then U is an upper bound of \mathcal{C} . So by Zorn's lemma \mathcal{O} has a maximal element.

3.3.5 Remark (a) By slight abuse of language we sometimes call an orthonormal family $(b_i)_{i \in I}$ in a Hilbert space \mathcal{H} an orthonormal basis or a Hilbert basis of \mathcal{H} if the set $\{b_i \mid i \in I\}$ is an orthonormal basis.

(b) If on an orthonormal basis $B \subset \mathcal{H}$ a total order relation is given, one calls B an ordered Hilbert basis of \mathcal{H} . Likewise, an orthonormal basis of the form $(b_i)_{i \in I}$ is called ordered if the index set I carries a total order.

3.3.6 Proposition (Pythagorean theorem for orthogonal families) An orthogonal family $(v_i)_{i \in I}$ in a Hilbert space \mathcal{H} is summable if and only if the family of norms $(||v_i||)_{i \in I}$ is square summable. In this case one has

$$\left\|\sum_{i\in I} v_i\right\|^2 = \sum_{i\in I} \|v_i\|^2 \ .$$

Proof. Assume that $(||v_i||)_{i \in I}$ is square summable or in other words that the net of partial sums $(\sum_{i \in J} ||v_i||^2)_{J \in \mathcal{P}_{fin}(I)}$ converges to some $s \in \mathbb{R}$. For $\varepsilon > 0$ choose a finite $J_{\varepsilon} \subset I$ such that for all finite J with $J_{\varepsilon} \subset J \subset I$ the relation

$$\left|s - \sum_{i \in J} \|v_i\|^2\right| < \frac{\varepsilon^2}{2}$$

holds true. For finite $K \subset I$ with $K \cap J_{\varepsilon} = \emptyset$ one then obtains by the pythagorean theorem for finite orthogonal families, Eq. (A.3.1.2),

$$\left\|\sum_{i\in K} v_i\right\|^2 = \sum_{i\in K} \|v_i\|^2 \leq \left|s - \sum_{i\in K\cup J_{\varepsilon}} \|v_i\|^2\right| + \left|s - \sum_{i\in J_{\varepsilon}} \|v_i\|^2\right| < \varepsilon^2$$

Hence $(\sum_{i \in J} v_i)_{J \in \mathcal{P}_{fin}(I)}$ is a Cauchy net in \mathcal{H} , so convergent.

Now let $(v_i)_{i \in I}$ be summable to $v \in \mathcal{H}$. Then there exists a $J_1 \in \mathcal{P}_{fin}(I)$ such that for all finite $J \subset I$ containing J_1

$$\left\|v - \sum_{i \in J} v_i\right\| \leq 1 \; .$$

This implies by the pythagorean theorem for finite orthogonal families

$$\sum_{i \in J} \|v_i\|^2 = \left\|\sum_{i \in J} v_i\right\|^2 \le \left(\left\|v - \sum_{i \in J} v_i\right\| + \|v\|\right)^2 \le (1 + \|v\|)^2.$$

Therefore, the net of partial sums $(\sum_{i \in J} ||v_i||^2)_{J \in \mathcal{P}_{fin}(I)}$ is bounded, so convergent since each term $||v_i||^2$ is non-negative.

By continuity of the inner product and pairwise orthogonality of the v_i one finally obtains in the convergent case

$$\left\|\sum_{i\in I} v_i\right\|^2 = \left\langle\sum_{i\in I} v_i, \sum_{j\in I} v_j\right\rangle = \sum_{i\in I} \left\langle v_i, \sum_{j\in I} v_j\right\rangle = \sum_{i\in I} \sum_{j\in I} \left\langle v_i, v_j\right\rangle = \sum_{i\in I} \|v_i\|^2 \quad .$$

3.3.7 Proposition Let $(v_i)_{i \in I}$ be an orthonormal family in a Hilbert space \mathcal{H} . Then for every $v \in \mathcal{H}$ the family $(\langle v_i, v \rangle)_{i \in I}$ is square summable and Bessel's inequality holds true that is

$$\sum_{i \in I} |\langle v_i, v \rangle|^2 \leqslant \|v\|^2 \ .$$

Proof. Let $J \subset I$ be finite. Then, by the pythagorean theorem for finite orthogonal families

$$0 \le \left\| v - \sum_{i \in J} \langle v_i, v \rangle v_i \right\|^2 = \|v\|^2 - 2\sum_{i \in J} |\langle v_i, v \rangle|^2 + \left\| \sum_{i \in J} \langle v_i, v \rangle v_i \right\|^2 = \|v\|^2 - \sum_{i \in J} |\langle v_i, v \rangle|^2 .$$

Therefore, for all $J \in \mathcal{P}_{fin}(I)$

0

$$\sum_{i \in J} |\langle v_i, v \rangle|^2 \leqslant ||v||^2 . \tag{A.3.3.1}$$

Hence, by Proposition 1.5.9, the family $(|\langle v_i, v \rangle|)_{i \in I}$ is square summable. Bessel's inequality now follows from the observation that in Equation (A.3.3.1) one can pass over to the limit of the net $\left(\sum_{i \in J} |\langle v_i, v \rangle|^2 \leq \|v\|^2\right)_{J \in \mathcal{P}_{c_n}(I)}$.

3.3.8 Theorem Let *B* be an orthonormal system in a Hilbert space \mathcal{H} . Then the following are equivalent:

- (1) The orthonormal system B is maximal, i.e. a Hilbert basis.
- (2) The orthonormal system B is total that is for all $v \in H$ such that $\langle v, b \rangle = 0$ for all $b \in B$ the equality v = 0 holds true.
- (3) For every $b \in B$ let $\mathcal{H}_b = \{rb \in \mathcal{H} \mid r \in \mathbb{K}\}$. Then the canonical map

$$\iota: \bigoplus_{b\in B} \mathcal{H}_b \to \mathcal{H}, \ (v_b)_{b\in B} \mapsto \sum_{b\in B} v_b$$

is an isometric isomorphism.

- (4) The closed linear span of B coincides with \mathcal{H} .
- (5) For all $v \in \mathcal{H}$, one has the Fourier expansion

$$v = \sum_{b \in B} \langle v, b \rangle b \; .$$

(6) For all $v, w \in \mathcal{H}$, one has

$$\langle v,w
angle = \sum_{b\in B} \langle v,b
angle\!\langle b,w
angle$$
 .

(7) For all $v \in \mathcal{H}$, Parseval's identity holds true that is

$$\|v\|^2 = \sum_{b \in B} |\langle v, b \rangle|^2 .$$

Proof. (1) \Rightarrow (2): If $v \neq 0$, then $\frac{v}{\|v\|}$ is a unit vector orthogonal to each v_i . Hence $\{v\} \cup B$ is an orthonormal system which is strictly larger than B, contradicting (1).

(2) \Rightarrow (3). First note that by the pythagorean theorem for infinite families, Proposition 3.3.6, the canonical map $\iota : \bigoplus_{b \in B} H_b \to H$ is well-defined and an isometry. Hence ι is injective. It remains to show that ι is surjective. To this end observe that $\operatorname{im} \iota$ is closed in \mathcal{H} since ι is an isometry (the image is complete). If ι is not surjective, then $\operatorname{im} \iota^{\perp}$ is not the zero vector space. Choose $v \in \operatorname{im} \iota^{\perp} \setminus \{0\}$. Then v is orthogonal to each element of B, but $v \neq 0$. This contradicts (2), so $\operatorname{im} \iota = \mathcal{H}$.

(3) \Rightarrow (5): We can represent any $v \in \mathcal{H}$ in the form $v = \iota((v_b)_{b\in B}) = \sum_{b\in B} v_b$ with $(v_b)_{b\in B} \in \widehat{\bigoplus}_{b\in B} H_b$. Write $v_b = r_b b$ for every $b \in B$, where $r_b \in \mathbb{K}$ is uniquely determined by v_b . Then compute using continuity of the inner product

$$\langle v, b \rangle = \langle \sum_{c \in B} v_c, b \rangle = \sum_{c \in B} r_c \langle c, b \rangle = r_b \; .$$

Therefore,

$$v = \sum_{b \in B} r_b \, b = \sum_{b \in B} \langle v, b \rangle b \; .$$

(5) \Rightarrow (6): Fourier expansion of $v, w \in H$ gives $v = \sum_{b \in B} \langle v, b \rangle b$ and $w = \sum_{b \in B} \langle w, b \rangle b$. Then, by continuity of the inner product,

$$\langle v, w \rangle = \sum_{b \in B} \langle v, b \rangle \langle b, w \rangle$$

(5) \Rightarrow (4): Let $v \in \mathcal{H}$. Then $\sum_{b \in J} \langle v, b \rangle b \in \operatorname{Span}(B)$ for all finite $J \subset B$. By Fourier expansion v is the limit of the net $\left(\sum_{b \in J} \langle v, b \rangle b\right)_{J \in \mathcal{P}_{\operatorname{fin}}(B)}$, so v lies in the closure $\overline{\operatorname{Span}}(B)$.

(4) \Rightarrow (2): Assume that $\langle v, b \rangle = 0$ for all $b \in B$. By (4), v can be written as a limit $v = \lim_{n \to \infty} v_n$, where $v_n \in \text{Span}(B)$ for all $n \in \mathbb{N}$. Then $\langle v, v_n \rangle = 0$ for all $n \in \mathbb{N}$ by assumption. By continuity of the inner product this implies

$$\|v\|^2 = \lim_{n \to \infty} \langle v, v_n \rangle = 0 ,$$

so v = 0.

(6) \Rightarrow (7): Put v = w. Then, by assumption,

$$\|v\|^2 = \langle v, v \rangle = \sum_{b \in B} \langle v, b \rangle \langle b, v \rangle = \sum_{b \in B} |\langle v, b \rangle|^2 \ .$$

(7) \Rightarrow (1): Assume (7) and that (1) is not true. Then there exists $v \in H$ with ||v|| = 1 and $\langle v, b \rangle = 0$ for all $b \in B$. But then

$$\|v\|^2 = \sum_{b \in B} |\langle v, b \rangle|^2 = 0,$$

which is a contradiction.

A.3.4. The monoidal structure of the category of Hilbert spaces

3.4.1 Let \mathbb{K} be the field of real or complex numbers. Hilbert spaces over \mathbb{K} together with bounded \mathbb{K} -linear maps between them form a category denoted by \mathbb{K} -Hilb or just Hilb if no confusion can arise. This can be seen immediately by observing that the identity map $\mathbb{1}_{\mathcal{H}}$ on a Hilbert space is a bounded linear operator and that the composition $B \circ A : \mathcal{H}_1 \to \mathcal{H}_3$ of two bounded linear operators between Hilbert spaces $A : \mathcal{H}_1 \to \mathcal{H}_2$ and $B : \mathcal{H}_2 \to \mathcal{H}_3$ is again a bounded linear operator. We want to endow the category Hilb with a bifunctor $\hat{\otimes}$: Hilb \times Hilb \to Hilb so that it becomes a monoidal category. The (bi)functor $\hat{\otimes}$ will be called the *Hilbert tensor product*.

Unless mentioned differently, Hilbert spaces, vector spaces and the algebraic tensor product \otimes in this section are assumed to be taken over the ground field \mathbb{K} .

3.4.2 Proposition Let \mathcal{H}_1 and \mathcal{H}_2 be two Hilbert spaces. Then there exists a unique inner product $\langle \cdot, \cdot \rangle : (\mathcal{H}_1 \otimes \mathcal{H}_2) \times (\mathcal{H}_1 \otimes \mathcal{H}_2) \rightarrow \mathbb{K}$ on the algebraic tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$ such that

$$\langle v_1 \otimes v_2, w_1 \otimes w_2 \rangle = \langle v_1, w_1 \rangle \cdot \langle v_2, w_2 \rangle \quad \text{for all } v_1, w_1 \in \mathcal{H}_1, \ v_2, w_2 \in \mathcal{H}_2 \ . \tag{A.3.4.1}$$

Proof. Let us first provide some preliminary constructions. Recall that for every pair of vector spaces V_1 and V_2 the bilinear map

$$\tau : \operatorname{Hom}(V_1, \mathbb{K}) \times \operatorname{Hom}(V_2, \mathbb{K}) \to \operatorname{Hom}(V_1 \otimes V_2, \mathbb{K}), (\lambda_1, \lambda_2) \mapsto (V_1 \otimes V_2 \to \mathbb{K}, v_1 \otimes v_2 \mapsto \lambda_1(v_1) \cdot \lambda_2(v_2))$$

induces a linear map

$$\hat{\tau} : \operatorname{Hom}(V_1, \mathbb{K}) \otimes \operatorname{Hom}(V_2, \mathbb{K}) \to \operatorname{Hom}(V_1 \otimes V_2, \mathbb{K})$$

by the universal property of the tensor product. This map is an isomorphism. To see this choose a basis $(v_{1i})_{i\in I}$ of V_1 and a basis $(v_{2j})_{j\in J}$ of V_2 . Let $(v'_{1i})_{i\in I}$ and $(v'_{2j})_{j\in J}$ denote the respective dual bases of V'_1 and V'_2 . Then $(v'_{1i} \otimes v'_{2j})_{(i,j)\in I \times J}$ is a basis of $\operatorname{Hom}(V_1, \mathbb{K}) \otimes \operatorname{Hom}(V_2, \mathbb{K})$ which under $\hat{\tau}$ is mapped bijectively to the basis $((v_{1i} \otimes v_{2j})')_{(i,j)\in I \times J}$ of $\operatorname{Hom}(V_1 \otimes V_2, \mathbb{K})$ dual to the basis $(v_{1i} \otimes v_{2j})_{(i,j)\in I \times J}$ of $V_1 \otimes V_2$. Hence $\hat{\tau}$ is a linear isomorphism as claimed, and we can identify the tensor product $\lambda_1 \otimes \lambda_2$ of two linear functionals $\lambda_i : V_i \to \mathbb{K}$, i = 1, 2 with its image in $\operatorname{Hom}(V_1 \otimes V_2, \mathbb{K})$.

Now observe that for two conjugate-linear maps $\mu_1 : V_1 \to \mathbb{K}$ and $\mu_2 : V_2 \to \mathbb{K}$ the map $\tau^*(\mu_1, \mu_2) = \overline{\overline{\mu_1} \otimes \overline{\mu_2}} : V_1 \otimes V_2 \to \mathbb{K}$ is conjugate-linear and satisfies

$$\tau^*(\mu_1,\mu_2) (v_1 \otimes v_2) = \mu_1(v_1) \cdot \mu_2(v_2) \quad \text{for all } v_1 \in \mathcal{V}_1, \ v_2 \in \mathcal{V}_2 \ . \tag{A.3.4.2}$$

One obtains a map

$$\tau^* : \operatorname{Hom}^*(V_1, \mathbb{K}) \times \operatorname{Hom}^*(V_2, \mathbb{K}) \to \operatorname{Hom}^*(V_1 \otimes V_2, \mathbb{K})$$

where here the symbol $\operatorname{Hom}^*(V, \mathbb{K})$ denotes the space of all conjugate linear functionals on a vector space V. Since τ^* is biadditive and since $\tau^*(z\mu_1, \mu_2) = \tau^*(\mu_1, z\mu_2)$ for all $\mu_1 \in \operatorname{Hom}^*(V_1, \mathbb{K})$, $\mu_2 \in \operatorname{Hom}^*(V_2, \mathbb{K})$, and $z \in \mathbb{K}$, the map τ^* factors through a linear map

$$\widehat{\tau^*}$$
: Hom^{*}(V₁, K) \otimes Hom^{*}(V₂, K) \rightarrow Hom^{*}(V₁ \otimes V₂, K)

Using the above bases $(v_{1i})_{i\in I}$ and $(v_{2j})_{j\in J}$ of V_1 and V_2 respectively, one observes that $\widehat{\tau^*}$ is an isomorphism since it maps the basis $(\overline{v'_{1i}} \otimes \overline{v'_{2j}})_{(i,j)\in I \times J}$ of $\operatorname{Hom}^*(V_1, \mathbb{K}) \otimes \operatorname{Hom}^*(V_2, \mathbb{K})$ bijectively to the basis $(\overline{(v_{1i} \otimes v_{2j})'})_{(i,j)\in I \times J}$ of the space $\operatorname{Hom}^*(V_1 \otimes V_2, \mathbb{K})$. So $\widehat{\tau^*}$ is also a linear isomorphism, which allows us to identify the tensor product $\mu_1 \otimes \mu_2$ of two conjugate linear functionals $\mu_i : V_i \to \mathbb{K}$, i = 1, 2 with its image in $\operatorname{Hom}^*(V_1 \otimes V_2, \mathbb{K})$.

After these preliminary considerations we consider the map

$$\mathcal{H}_1 \times \mathcal{H}_2 \to \operatorname{Hom}^*(\mathcal{H}_1 \otimes \mathcal{H}_2, \mathbb{K}), \ (w_1, w_2) \mapsto \overline{w_1^{\flat}} \otimes \overline{w_2^{\flat}} = \tau^* \left(\overline{w_1^{\flat}}, \overline{w_2^{\flat}} \right) = \widehat{\tau^*} \left(\overline{w_1^{\flat}} \otimes \overline{w_2^{\flat}} \right)$$

which is well-defined and bilinear since the musical isomorphisms ${}^{\flat} : \mathcal{H}_l \to \mathcal{H}'_l$, $w \mapsto \langle w, - \rangle$, l = 1, 2, are conjugate-linear and since τ^* is bilinear. Hence it factors through a linear map

$$\beta: \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathrm{Hom}^*(\mathcal{H}_1 \otimes \mathcal{H}_2, \mathbb{K})$$

such that

$$\beta(w_1 \otimes w_2)(v_1 \otimes v_2) = \langle v_1, w_1 \rangle \cdot \langle v_2, w_2 \rangle \quad \text{for all } v_1, w_1 \in \mathcal{H}_1, \ v_2, w_2 \in \mathcal{H}_2 \ . \tag{A.3.4.3}$$

Now put

$$\langle \cdot, \cdot \rangle : (\mathcal{H}_1 \otimes \mathcal{H}_2) \times (\mathcal{H}_1 \otimes \mathcal{H}_2) \to \mathbb{K}, \ (v, w) \mapsto \langle v, w \rangle := \beta(w)(v) \ .$$

Then $\langle \cdot, \cdot \rangle$ is sesquilinear by construction, and (A.3.4.1) holds true by (A.3.4.3).

Let us show that $\langle \cdot, \cdot \rangle$ is positive definite. Let $v = \sum_{k=1}^{n} v_{1k} \otimes v_{2k} \in \mathcal{H}_1 \otimes \mathcal{H}_2$. Choose an orthonormal basis e_1, \ldots, e_m of the linear subspace spanned by v_{21}, \ldots, v_{2n} . Expand $v_{2k} = \sum_{i=1}^{m} c_{ki}e_i$ with $c_{k1}, \ldots, c_{km} \in \mathbb{K}$. Then

$$v = \sum_{k=1}^{n} v_{1k} \otimes v_{2k} = \sum_{k=1}^{n} \sum_{i=1}^{m} v_{1k} \otimes (c_{ki}e_i) = \sum_{i=1}^{m} \left(\sum_{k=1}^{n} c_{ki}v_{1k}\right) \otimes e_i = \sum_{i=1}^{m} w_{1i} \otimes e_i , \quad (A.3.4.4)$$

where $w_{1i} = \sum_{k=1}^{n} c_{ki} v_{1k}$. Hence

<

$$\langle v, v \rangle = \langle \sum_{i=1}^{m} w_{1i} \otimes e_i, \sum_{j=1}^{m} w_{1j} \otimes e_j \rangle = \sum_{i=1}^{m} \sum_{j=1}^{m} \langle w_{1i}, w_{1j} \rangle \langle e_i, e_j \rangle = \sum_{i=1}^{m} \|w_{1i}\|^2 \ge 0 .$$
 (A.3.4.5)

Moreover, if $\langle v, v \rangle = 0$, then $w_{1i} = 0$ for i = 1, ..., m, which implies $v = \sum_{i=1}^{m} w_{1i} \otimes e_i = 0$. So $\langle \cdot, \cdot \rangle$ is an inner product on $\mathcal{H}_1 \otimes \mathcal{H}_2$ satisfying (A.3.4.1). It is uniquely determined by this condition since the vectors $v_1 \otimes v_2$ with $v_1 \in \mathcal{H}_1$ and $v_2 \in \mathcal{H}_2$ span $\mathcal{H}_1 \otimes \mathcal{H}_2$.

3.4.3 Definition Let \mathcal{H}_1 and \mathcal{H}_2 be Hilbert spaces. The Hilbert completion of the algebraic tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$ equipped with the unique inner product $\langle \cdot, \cdot \rangle$ fulfilling (A.3.4.1) will be denoted $\mathcal{H}_1 \otimes \mathcal{H}_2$, its inner product again by $\langle \cdot, \cdot \rangle$. One calls the Hilbert space $(\mathcal{H}_1 \otimes \mathcal{H}_2, \langle \cdot, \cdot \rangle)$ the *Hilbert tensor product* of \mathcal{H}_1 and \mathcal{H}_2 or just the *tensor product* of \mathcal{H}_1 and \mathcal{H}_2 if no confusion can arise.

3.4.4 Proposition Let \mathcal{H}_1 and \mathcal{H}_2 be Hilbert spaces.

- (i) Let $A_1 \subset \mathfrak{H}_1$ and $A_2 \subset \mathfrak{H}_2$ be subsets which are total \mathfrak{H}_1 and \mathfrak{H}_2 , respectively. Then the set of simple vectors $a_1 \otimes a_2$ with $a_1 \in A_1$ and $a_2 \in A_2$ is total in the Hilbert tensor product $\mathfrak{H}_1 \otimes \mathfrak{H}_2$.
- (ii) If $(e_i)_{i \in I}$ and $(f_j)_{j \in J}$ are orthonormal bases of \mathcal{H}_1 and \mathcal{H}_2 , respectively, then $(e_i \otimes f_j)_{(i,j) \in I \times J}$ is an orthonormal basis of the Hilbert tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$.

Proof. ad (i). Recall that a subset $A \subset \mathcal{H}$ or a family $A = (a_j)_{j \in J}$ of elements of a Hilbert space \mathcal{H} is called *total* in \mathcal{H} if the linear span of A is dense in \mathcal{H} . By density of the algebraic tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$ in the Hilbert tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$, the set of simple tensors $v_1 \otimes v_2$ with $(v_1, v_2) \in \mathcal{H}_1 \times \mathcal{H}_2$ is total in $\mathcal{H}_1 \otimes \mathcal{H}_2$. Hence it suffices to find for each such pair (v_1, v_2) and all $\varepsilon > 0$ vectors $w_1 \in \operatorname{Span} A_1$ and $w_2 \in \operatorname{Span} A_2$ such that

$$\|v_1 \otimes v_2 - w_1 \otimes w_2\| < \frac{\varepsilon}{2} .$$

By totality of A_i in \mathcal{H}_i there exist $w_i \in \operatorname{Span} A_i$ for i = 1, 2 such that

$$||v_1 - w_1|| < \min\left\{1, \frac{\varepsilon}{2(||v_2|| + 1)}\right\}$$
 and $||v_2 - w_2|| < \frac{\varepsilon}{2(||v_1|| + 1)}$.

Then

$$||v_1 \otimes v_2 - w_1 \otimes w_2|| \le ||v_1 - w_1|| ||v_2|| + ||v_2 - w_2||||w_1|| < \varepsilon$$
.

ad (ii). The family $(e_i \otimes f_j)_{(i,j) \in I \times J}$ is orthonormal by definition of the inner product on $\mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$. It is total by (i) and therefore a Hilbert basis.

3.4.5 Proposition Assigning to each pair of Hilbert spaces \mathcal{H}_1 and \mathcal{H}_2 the Hilbert tensor product $\mathcal{H}_1 \otimes \mathcal{H}_2$ and to each pair of bounded linear operators $A_1 : \mathcal{H}_1 \to \mathcal{H}_3$ and $A_2 : \mathcal{H}_2 \to \mathcal{H}_4$ between Hilbert spaces the unique bounded extension $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator of $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2 \to \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_1 \otimes \mathcal{H}_2$ operator $A_1 \otimes \mathcal{H}_2$ operator $A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_2 \otimes \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes A_2 : \mathcal{H}_3 \otimes \mathcal{H}_4$ of the operator $A_1 \otimes \mathcal{H}_4$ of the operator

 $\widehat{\otimes}\,: \mathsf{Hilb} \times \mathsf{Hilb} \to \mathsf{Hilb}$.

Moreover, $\hat{\otimes}$ is isometric in the sense that

$$||v_1 \otimes v_2|| = ||v_1|| ||v_2||$$
 for all $v_1 \in \mathcal{H}_1, v_2 \in \mathcal{H}_1$ and (A.3.4.6)

$$\|A_1 \widehat{\otimes} A_2\| = \|A_1\| \, \|A_2\| \quad \text{for all } A_1 \in \mathfrak{B}(\mathfrak{H}_1, \mathfrak{H}_3), \, A_2 \in \mathfrak{B}(\mathfrak{H}_2, \mathfrak{H}_4) \,. \tag{A.3.4.7}$$

Proof. We first show that $A_1 \otimes A_2$ is a bounded operator. To this end observe that $A_1 \otimes A_2$ can be written as the composition of the two operators $A_1 \otimes \mathbb{1}_{\mathcal{H}_2}$ and $\mathbb{1}_{\mathcal{H}_3} \otimes A_2$. Hence it suffices to show that each of these linear maps is bounded. Let $v = \sum_{k=1}^n v_{1k} \otimes v_{2k} \in \mathcal{H}_1 \otimes \mathcal{H}_2$ be of norm 1. As in the proof of Proposition 3.4.2 expand $v_{2k} = \sum_{i=1}^m c_{ki}e_i$, $k = 1, \ldots, n$, where e_1, \ldots, e_m is an orthonormal basis of $\operatorname{Span}\{v_{21}, \ldots, v_{2n}\}$ and $c_{k1}, \ldots, c_{km} \in \mathbb{K}$. Equations (A.3.4.4) and (A.3.4.5) then entail that

$$v = \sum_{i=1}^{m} w_{1i} \otimes e_i \quad \text{and} \quad 1 = \langle v, v \rangle = \sum_{i=1}^{m} \|w_{1i}\|^2$$

where $w_{1i} = \sum_{k=1}^{n} c_{ki} v_{1k}$ for $i = 1, \ldots, m$. Hence

$$\|(A_1 \otimes \mathbb{1}_{\mathcal{H}_2})v\|^2 = \left\|\sum_{i=1}^m A_1(w_{1i}) \otimes e_i\right\|^2 = \sum_{i=1}^m \|A_1(w_{1i})\|^2 \le \|A_1\|^2 \sum_{i=1}^m \|w_{1i}\|^2 = \|A_1\|^2,$$

so $A_1 \otimes \mathbb{1}_{\mathcal{H}_2}$ is bounded with norm $\leq ||A_1||$. By symmetry, $\mathbb{1}_{\mathcal{H}_3} \otimes A_2$ is bounded with norm $\leq ||A_2||$. Hence $A_1 \otimes A_2 = (\mathbb{1}_{\mathcal{H}_3} \otimes A_2) \circ (A_1 \otimes \mathbb{1}_{\mathcal{H}_2})$ is bounded and

$$||A_1 \otimes A_2|| \leq ||A_1|| ||A_2||$$
.

Therefore, $A_1 \otimes A_2$ has a unique bounded extension $A_1 \otimes A_2$ of norm

$$|A_1 \otimes A_2|| = ||A_1 \otimes A_2|| \le ||A_1|| \, ||A_2||$$

Let us show that the converse inequality holds as well. For given $\varepsilon > 0$ there exist unit vectors $v_i \in \mathcal{H}_i$, i = 1, 2 such that $||A_i v_i|| \ge ||A_i|| - \frac{\varepsilon}{2(||A_1|| + ||A_2|| + 1)}$. Then

$$||(A_1 \otimes A_2)(v_1 \otimes v_2)|| = ||A_1v_1|| \, ||A_2v_2|| \ge ||A_1|| \, ||A_2|| - \varepsilon \, .$$

This implies

$$||A_1 \otimes A_2|| = ||A_1 \otimes A_2|| \ge ||A_1|| ||A_2||$$

and (A.3.4.7) follows. Equality (A.3.4.6) is clear by construction of the Hilbert tensor product.

Next observe that $\mathbb{1}_{\mathcal{H}_1} \otimes \mathbb{1}_{\mathcal{H}_2} = \mathbb{1}_{\mathcal{H}_1 \otimes \mathcal{H}_2}$ by definition. Given Hilbert spaces $\mathcal{H}_1, \ldots, \mathcal{H}_6$ and bounded linear operators $A_i : \mathcal{H}_i \to \mathcal{H}_{i+2}$ and $B_i : \mathcal{H}_{i+2} \to \mathcal{H}_{i+4}$ for i = 1, 2, the composition $(B_1 \otimes B_2) \circ$

 $(A_1 \otimes A_2)$ coincides with $(B_1 \circ A_1) \otimes (B_2 \circ A_2)$ by functoriality of the algebraic tensor product. By continuity of the operators $A_1 \otimes A_2$ and $B_1 \otimes B_2$ and by density of $\mathcal{H}_1 \otimes \mathcal{H}_2$ in $\mathcal{H}_1 \otimes \mathcal{H}_2$ the equality

$$(B_1 \widehat{\otimes} B_2) \circ (A_1 \widehat{\otimes} A_2) = (B_1 \circ A_1) \widehat{\otimes} (B_2 \circ A_2)$$

follows. Hence $\hat{\otimes}$ is a bifunctor as claimed.

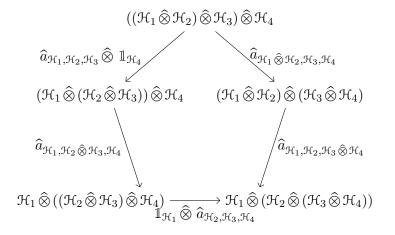
3.4.6 Proposition For every Hilbert space \mathcal{H} one has two natural isomorphisms

$$\hat{u}_{\mathcal{H}}: \mathbb{K} \widehat{\otimes} \mathcal{H} \to \mathcal{H}, \ z \otimes v \to zv \quad \text{and} \quad_{\mathcal{H}} \hat{u}: \mathcal{H} \widehat{\otimes} \mathbb{K} \to \mathcal{H}, \ v \otimes z \to zv$$

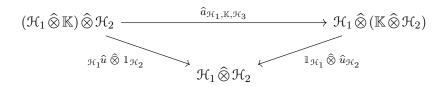
called the left and the right unit, respectively. Furthermore, for every triple of Hilbert spaces $\mathcal{H}_1, \mathcal{H}_2, \mathcal{H}_3$ there is a natural isomorphism, called associator

 $\hat{a}_{\mathcal{H}_1,\mathcal{H}_2,\mathcal{H}_3}: (\mathcal{H}_1 \,\widehat{\otimes}\, \mathcal{H}_2) \,\widehat{\otimes}\, \mathcal{H}_3 \to \mathcal{H}_1 \,\widehat{\otimes}\, (\mathcal{H}_2 \,\widehat{\otimes}\, \mathcal{H}_3), \ (v_1 \otimes v_2) \otimes v_3 \mapsto v_1 \otimes (v_2 \otimes v_3) \ .$

These data fulfill the so-called coherence conditions that is the pentagon diagram



and the triangle diagram



commute for all Hilbert spaces $\mathcal{H}_1, \mathcal{H}_2, \mathcal{H}_3, \mathcal{H}_4$. In other words, the category Hilb endowed with the Hilbert tensor product $\hat{\otimes}$ is a monoidal category.

Proof. The category of K-vector spaces with the usual tensor product as tensor functor is monoidal. Denote the corresponding unit isomorphisms and associator by $_{-u}$, u_{-} , and $a_{-,-,-}$, respectively. Then observe that by construction $\mathbb{K}\widehat{\otimes}\mathcal{H} = \mathbb{K}\otimes\mathcal{H}$ and $\mathcal{H}\widehat{\otimes}\mathbb{K} = \mathcal{H}\otimes\mathbb{K}$ for every Hilbert space \mathcal{H} . In particular this means that $\hat{u}_{\mathcal{H}}$ coincides with the unit $u_{\mathcal{H}}$ and $_{\mathcal{H}}\hat{u}$ with the unit $_{\mathcal{H}}u$. Moreover, both units $\hat{u}_{\mathcal{H}}$ and $_{\mathcal{H}}\hat{u}$ are bounded. Next recall that $\mathcal{H}_1 \otimes \mathcal{H}_2$ is dense in $\mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2$ which by Proposition 3.4.4 implies density of $(\mathcal{H}_1 \otimes \mathcal{H}_2) \otimes \mathcal{H}_3$ and $\mathcal{H}_1 \otimes (\mathcal{H}_2 \otimes \mathcal{H}_3)$ in $(\mathcal{H}_1 \widehat{\otimes} \mathcal{H}_2) \widehat{\otimes} \mathcal{H}_3$ and $\mathcal{H}_1 \widehat{\otimes} (\mathcal{H}_2 \widehat{\otimes} \mathcal{H}_3)$, respectively. Similarly one argues that $\mathcal{H}_1 \otimes (\mathcal{H}_2 \otimes (\mathcal{H}_3 \otimes \mathcal{H}_4))$ is dense in

 $\mathcal{H}_1 \,\widehat{\otimes}\, (\mathcal{H}_2 \,\widehat{\otimes}\, (\mathcal{H}_3 \,\widehat{\otimes}\, \mathcal{H}_4))$, and so on. Since the associator map $a_{\mathcal{H}_1, \mathcal{H}_2, \mathcal{H}_3} : (\mathcal{H}_1 \otimes \mathcal{H}_2) \otimes \mathcal{H}_3 \rightarrow \mathcal{H}_1 \otimes (\mathcal{H}_2 \otimes \mathcal{H}_3)$ is bounded, it extends in a unique way to a linear bounded map $\hat{a}_{\mathcal{H}_1, \mathcal{H}_2, \mathcal{H}_3} : (\mathcal{H}_1 \,\widehat{\otimes}\, \mathcal{H}_2) \,\widehat{\otimes}\, \mathcal{H}_3 \rightarrow \mathcal{H}_1 \,\widehat{\otimes}\, (\mathcal{H}_2 \,\widehat{\otimes}\, \mathcal{H}_3)$. Using density, continuity, and commutativity of the pentagon and triangle diagrams for the tensor product functor one concludes that the coherence conditions for $\widehat{\otimes}$ with the unit and associator maps $_{-}\hat{u}, \, \hat{u}_{-}$, and $\hat{a}_{-,-,-}$ are satisfied.

A.3.5. Adjoints of bounded operators

3.5.1 As before, the symbols \mathcal{H} and \mathcal{H}_k with k = 1, 2 always stand for Hilbert spaces over the field \mathbb{K} of real or complex numbers. Several results of this section hold only in the complex case, thouhgh. Therefore we will be quite precise in stating all necessary assumptions, in particular about the ground field.

Let $A \in \mathfrak{B}(\mathfrak{H}_1, \mathfrak{H}_2)$ that is let $A : \mathfrak{H}_1 \to \mathfrak{H}_2$ be linear and bounded. Then the map

$$b_A: \mathcal{H}_1 \times \mathcal{H}_2 \to \mathbb{K}, \ (v, w) \mapsto \langle Av, w \rangle$$

is sesquilinear and bounded with norm

$$\|b_A\| = \sup\{ |b_A(v, w)| \mid v \in \mathcal{H}_1, w \in \mathcal{H}_2, \|w\| = \|v\| = 1 \} = \|A\| .$$

By Corollary 3.2.8 to the Riesz representation theorem there exists a unique bounded linear operator $A^* : \mathcal{H}_2 \to \mathcal{H}_1$ such that

$$b_A(v,w) = \langle v, A^*w \rangle$$
 for all $v \in \mathcal{H}_1, w \in \mathcal{H}_2$.

This operator satisfies

$$||A^*|| = ||b_A|| = ||A|| . (A.3.5.1)$$

3.5.2 Definition The unique operator $A^* \in \mathfrak{B}(\mathfrak{H}_2, \mathfrak{H}_1)$ associated to an operator $A \in \mathfrak{B}(\mathfrak{H}_1, \mathfrak{H}_2)$ such that

 $\langle Av, w \rangle = \langle v, A^*w \rangle$ for all $v \in \mathcal{H}_1, w \in \mathcal{H}_2$

is called the *adjoint* of A.

The fundamental property of the adjoint operation is given by the following result.

3.5.3 Proposition The adjoint map $* : \mathfrak{B}(\mathfrak{H}_1, \mathfrak{H}_2) \to \mathfrak{B}(\mathfrak{H}_2, \mathfrak{H}_1)$ is a conjugate linear isometry whose square coincides with the identity operation that is $A^{**} = A$ for all $A \in \mathfrak{B}(\mathfrak{H}_1, \mathfrak{H}_2)$.

Proof. By the proof of Corollary 3.2.8, $A^*w = \langle w, A(-) \rangle^{\sharp}$ for all $w \in \mathcal{H}_2$. Since the inner product is linear in the second argument and the operator \sharp conjugate linear, the map $A \mapsto A^*$ is conjugate linear in A. By Equation (A.3.5.1), the adjoint map is an isometry. The relation $A^{**} = A$ follows by uniqueness of the adjoint and since

$$\langle A^*w, v \rangle = \langle w, Av \rangle$$
 for all $v \in \mathcal{H}_1, w \in \mathcal{H}_2$.

3.5.4 Definition An operator $A \in \mathfrak{B}(\mathcal{H})$ is called *self-adjoint* if $A = A^*$, *unitary* if $A^* = A^{-1}$, and *normal* if $[A, A^*] := AA^* - A^*A = 0$.

We note that self-adjoint and unitary operators are always normal, but normal operators do not have to be self-adjoint or unitary. In the remainder of this section, we gather several results on self-adjoint and normal operators.

3.5.5 Proposition Assume that the ground field \mathbb{K} of the Hilbert space \mathcal{H} is the field of complex numbers. An operator $A \in \mathfrak{B}(\mathcal{H})$ then is self-adjoint if and only if $\langle Av, v \rangle \in \mathbb{R}$ for all $v \in \mathcal{H}$.

Proof. (\Rightarrow) If A is self-adjoint, then

$$\langle Av, v \rangle = \langle v, A^*v \rangle = \langle v, Av \rangle = \overline{\langle Av, v \rangle},$$

which implies that $\langle Av, v \rangle \in \mathbb{R}$.

(\Leftarrow) Suppose that $\langle Av, v \rangle \in \mathbb{R}$ for all $v \in \mathcal{H}$. We know

$$\langle A(v+w), v+w \rangle = \langle Av, v \rangle + \langle Av, w \rangle + \langle Aw, v \rangle + \langle Aw, w \rangle .$$
(A.3.5.2)

By assumption, $\langle A(v+w), v+w \rangle$, $\langle Av, v \rangle$, and $\langle Aw, w \rangle$ are all real. This implies that the sum $\langle Av, w \rangle + \langle Aw, v \rangle$ is real as well, so

$$\Im\mathfrak{m}\langle Av,w\rangle = -\Im\mathfrak{m}\langle Aw,v\rangle = \Im\mathfrak{m}\langle v,Aw\rangle \ .$$

Since this holds for all $w \in \mathcal{H}$, it holds for iw, too. Thus,

$$\mathfrak{Re}\langle Av, w \rangle = \mathfrak{Im} \operatorname{i} \langle Av, w \rangle = \mathfrak{Im} \langle Av, \mathsf{i} w \rangle = \mathfrak{Im} \langle v, A(\mathsf{i} w) \rangle = \mathfrak{Im} \operatorname{i} \langle v, Aw \rangle = \mathfrak{Re} \langle v, Aw \rangle.$$

Combining the above two lines yields $\langle Av, w \rangle = \langle v, Aw \rangle$ for all $v, w \in \mathcal{H}$. By uniqueness of the adjoint this implies that $A = A^*$.

3.5.6 Proposition Assume that the ground field \mathbb{K} of the Hilbert space \mathcal{H} is the field of complex numbers and let $A \in \mathfrak{B}(\mathcal{H})$. If $\langle Av, v \rangle = 0$ holds for all $v \in \mathcal{H}$, then A = 0.

Proof. Since $\langle Av, v \rangle = 0$ for all $v \in H$, equation (A.3.5.2) from the proof of Proposition 3.5.5 reduces to

$$Av, w
angle = -\langle Aw, v
angle = -\langle w, Av
angle = -\overline{\langle Av, w
angle} \quad \text{for all } v, w \in \mathfrak{H} \; .$$

That means that $\langle Av, w \rangle$ has no real part for all $v, w \in \mathcal{H}$. But then fixing v and setting w = Av implies $||Av||^2 = 0$ for all $v \in \mathcal{H}$, so A = 0.

3.5.7 Example The preceding proposition does not hold in the real case. To see this take rotation by $\frac{\pi}{2}$:

$$R = \begin{pmatrix} \cos\frac{\pi}{2} & -\sin\frac{\pi}{2} \\ \sin\frac{\pi}{2} & \cos\frac{\pi}{2} \end{pmatrix}$$

Then $\langle Rv, v \rangle = 0$ for all $v \in \mathbb{R}^2$, but R is non-zero. Note that the example of the rotation operator R also shows that the criterion for self-adjointness from Proposition 3.5.5 can not be applied in the real case.

3.5.8 Lemma (cf. (Hirzebruch & Scharlau, 1991, Lem. 22.4)) Assume that A is a bounded linear operator on the real or complex Hilbert space \mathcal{H} for which there exists a $C \ge 0$ such that

$$|\langle Av, v \rangle| \leq C ||v||^2$$
 for all $v \in \mathcal{H}$.

Then

$$\langle Av, w \rangle + \langle v, Aw \rangle | \leq 2C ||v|| ||w|| \quad \text{for all } v, w \in \mathcal{H}$$
. (A.3.5.3)

In case $\mathcal H$ is a complex Hilbert space one even has the sharper estimate

$$|\langle Av, w \rangle| + |\langle v, Aw \rangle| \leq 2C ||v|| ||w|| \quad \text{for all } v, w \in \mathcal{H} .$$
(A.3.5.4)

Proof. We start with the equality

$$\langle A(v+w), v+w \rangle + \langle A(v-w), v-w \rangle = 2(\langle Av, w \rangle + \langle Aw, v \rangle) .$$
(A.3.5.5)

By assumption and the parallelogram identity (A.3.1.3) this entails

$$2|\langle Av, w \rangle + \langle Aw, v \rangle| \le C \left(\|v + w\|^2 + \|v - w\|^2 \right) = 2C \left(\|v\|^2 + \|w\|^2 \right) .$$
(A.3.5.6)

The claim obviously holds for v = 0 or w = 0, so we assume from now on that both v and w are non-zero. Then put $a = \sqrt{\frac{\|v\|}{\|w\|}}$ and replace in (A.3.5.6) v by $\frac{v}{a}$ and w by aw. One obtains

$$\left|\left\langle Av,w\right\rangle +\left\langle Aw,v\right\rangle\right|\leqslant C\left(\left\|\frac{v}{a}\right\|^{2}+\left\|aw\right\|^{2}\right)=2C\|v\|\|w\|$$

which is the claim in the real case. If \mathcal{H} is a complex Hilbert space, let x, y be complex numbers of modulus 1. In the just proven estimate multiply the left side with |x| and replace w with yw. This gives

$$|xy\langle Av, w\rangle + x\overline{y}\langle Aw, v\rangle| = |x| \cdot |\langle Av, yw\rangle + \langle A(yw), v\rangle| \le 2C ||v|| ||w|| .$$
(A.3.5.7)

Now write $\langle Av, w \rangle = r e^{i\varphi}$ and $\langle Aw, v \rangle = s e^{i\psi}$ with $r, s \ge 0$ and $\varphi, \psi \in \mathbb{R}$. Then put

$$x = e^{-i\frac{1}{2}(\varphi+\psi)}$$
 and $y = e^{-i\frac{1}{2}(\varphi-\psi)}$

With these values, (A.3.5.7) becomes

$$|\langle Av, w \rangle| + |\langle v, Aw \rangle| \leqslant 2C \|v\| \|w\|$$

which was to be shown.

3.5.9 Proposition If \mathcal{H} is a Hilbert space over the field \mathbb{K} of real or complex numbers and $A \in \mathfrak{B}(\mathcal{H})$ is self-adjoint, then

$$||A|| = \sup_{||v||=1} |\langle Av, v \rangle|$$

Proof. We know

$$||A|| = \sup_{\|v\| = \|w\| = 1} |\langle Av, w \rangle| , \qquad (A.3.5.8)$$

so we clearly have

$$\sup_{\|v\|=1} |\langle Av, v \rangle| \le \|A\|$$

The other direction follows from Equation (A.3.5.8) and Lemma 3.5.8 since A is self-adjoint. \Box

3.5.10 Proposition If \mathcal{H} is a real or complex Hilbert space and $A \in \mathfrak{B}(\mathcal{H})$, then A^*A is self-adjoint and $||A^*A|| = ||A||^2$.

Proof. For arbitrary $v, w \in \mathcal{H}$, we have

$$\langle A^*Av, w \rangle = \langle Av, Aw \rangle = \langle v, A^*Aw \rangle$$

so A^*A is self-adjoint. Then

$$||A^*A|| = \sup_{\|v\| = \|w\| = 1} |\langle A^*Av, w \rangle| = \sup_{\|v\| = \|w\| = 1} |\langle Av, Aw \rangle| = ||A||^2 ,$$

where the last equality is a consequence of the Cauchy–Schwarz inequality and the observation that for all $\varepsilon > 0$ there exists a unit vector v such that $\langle Av, Av \rangle \ge ||A||^2 - \varepsilon$.

3.5.11 Proposition Let \mathcal{H} be a complex Hilbert space \mathcal{H} . If $A \in \mathfrak{B}(\mathcal{H})$, then there exist unique self-adjoint $B, C \in \mathfrak{B}(\mathcal{H})$ such that A = B + iC. Furthermore, A is normal if and only if [B, C] = 0.

Proof. We define

$$B = \frac{1}{2}(A + A^*)$$
 and $C = \frac{i}{2}(A^* - A).$

Clearly A = B + iC. Note also that $A^* = B - iC$. Furthermore, by Proposition 3.5.3

$$B^* = \frac{1}{2}(A^* + A) = B$$

and

$$C^* = -\frac{i}{2}(A - A^*) = C$$
.

Hence B and C are self-adjoint, so fulfill the claim. Let us show uniqueness. Assume that $B', C' \in \mathfrak{B}(\mathfrak{H})$ are selfadjoint and satisfy A = B' + iC'. Then

$$B - B' = B^* - B'^* = (i(C' - C))^* = -i(C' - C) = -(B - B').$$

Hence B = B' and consequently C = C'. Finally, we compute

$$[A, A^*] = [B + iC, B - iC] = -i[B, C] + i[C, B] = -2i[B, C] .$$

This entails that A is normal if and only if [B, C] = 0.

3.5.12 Proposition If A is a normal operator on a real or complex Hilbert space \mathcal{H} , then

$$||Av|| = ||A^*v|| \quad \text{for all } v \in \mathcal{H}$$

Proof. Using the fact that $A^*A = AA^*$, we compute

$$Av\|^{2} = \langle Av, Av \rangle = \langle v, A^{*}Av \rangle = \langle v, AA^{*}v \rangle = \langle A^{*}v, A^{*}v \rangle = \|A^{*}v\|^{2}$$

Taking the square root yields the desired result.

A.3.6. Projection-valued measures and spectral integrals

3.6.1 In this section $\mathcal H$ will always denote a fixed complex Hilbert space.

3.6.2 Definition By a *projection-valued measure* or a *spectral measure* on a measurable space (Ω, \mathcal{A}) one understands a map $E : \mathcal{A} \to \mathfrak{B}(\mathcal{H})$ having the following properties:

- (SM0) For each $\Delta \in \mathcal{A}$ the operator $E(\Delta)$ is an orthogonal projection that is $E(\Delta)^2 = E(\Delta)$ and $E(\Delta)^* = E(\Delta)$.
- (SM1) $E(\Omega) = \mathrm{id}_{\mathcal{H}}.$
- (SM2) For every sequence $(\Delta_n)_{n\in\mathbb{N}}$ of pairwise disjoint elements of \mathcal{A} one has

$$E\left(\bigcup_{n\in\mathbb{N}}\Delta_n\right) = \mathbf{s} \cdot \sum_{n=0}^{\infty} E(\Delta_n) ,$$

where convergence is with respect to the strong operator toplogy.

3.6.3 Remark Recall that *convergence* of a sequence of operators $(A_n)_{n \in \mathbb{N}} \subset \mathfrak{B}(\mathcal{H})$ in the *strong* operator topology to some A means that for every $v \in \mathcal{H}$ the sequence $(A_n v)_{n \in \mathbb{N}}$ converges in \mathcal{H} to Av. One denotes this by $A = \operatorname{s-lim}_{n \to \infty} A_n$. Likewise, $B = \operatorname{s-s}_{n=0}^{\infty} A_n$ means that the sequence of partial $\operatorname{s-s}_{n=0}^{\infty} A_n$ means that the sequence of partial $\operatorname{s-s}_{n=0}^{\infty} A_n$.

sums $\left(\sum_{k=0}^{n} A_{n}\right)_{n \in \mathbb{N}}$ converges in the strong operator topology to some $B \in \mathfrak{B}(\mathcal{H})$.

3.6.4 Proposition A spectral measure $E : \mathcal{A} \to \mathfrak{B}(\mathfrak{H})$ has the following properties in addition to the defining axioms:

 $(\mathsf{SM1'}) E(\emptyset) = 0.$

(SM2') (Finite additivity) One has for all disjoint $\Delta_1, \Delta_2 \in \mathcal{A}$

$$E(\Delta_1 \cup \Delta_2) = E(\Delta_1) + E(\Delta_2)$$
.

(SM3) One has for all $\Delta_1, \Delta_2 \in \mathcal{A}$

$$E(\Delta_1 \cap \Delta_2) = E(\Delta_1) \cdot E(\Delta_2)$$
.

Proof. ad (SM1').

ad (SM2').

ad (SM3).

A.3.7. Spectral theory of bounded operators

3.7.1 We now apply the foundations of Hilbert space theory built in the previous sections to spectral theory. For the moment we will sacrifice generality and work only with bounded linear operators. The spectral theory of unbounded linear operators will be treated later.

Let us a recall that a linear map $A : \mathcal{H}_1 \to \mathcal{H}_2$ between Hilbert spaces is continuous if and only if it is bounded, i.e. has finite operator norm, and that $\mathfrak{B}(\mathcal{H}_1, \mathcal{H}_2)$ is a Banach space with the operator norm. For the rest of this section, $\mathcal{H}, \mathcal{H}_1, \mathcal{H}_2, \ldots$ will always denote complex Hilbert spaces and A, B bounded linear operators. We will also now fix the base field to be complex, i.e. $\mathbb{K} = \mathbb{C}$. Last we agree on writing $I_{\mathcal{H}}$ or just I for the identity operator on a Hilbert space \mathcal{H} .

Spectrum and Resolvent

3.7.2 Definition Let $A : \mathcal{H} \to \mathcal{H}$ be a bounded linear operator. A complex number λ is then called an *eigenvalue* of A if there exists a nonzero $v \in H$ such that $Av = \lambda v$. For every $\lambda \in \mathbb{C}$ one defines the λ -*eigenspace* of A as

$$\operatorname{Eig}_{\lambda}(A) = \{ v \in H \mid Av = \lambda v \} \subset \mathcal{H},$$

which is clearly a linear subspace of \mathcal{H} .

3.7.3 By definition it is immediately clear that

$$\operatorname{Eig}_{\lambda}(A) = \ker(A - \lambda),$$

where the λ on the right stands for the operator λI . In other words this means that $\lambda \in \mathbb{C}$ is an eigenvalue of A if and only if $A - \lambda$ is not injective.

3.7.4 Definition Let $A \in \mathfrak{B}(\mathcal{H})$. We make the following definitions.

- (i) A regular value of A is a complex number λ such that $A \lambda$ is invertible.
- (ii) The set of all regular values is the *resolvent* of A, denoted $\rho(A)$.
- (iii) A spectral value of A is a complex number λ such that $A \lambda$ is not invertible.
- (iv) The set of all spectral values is the *spectrum* of A, denoted $\sigma(A)$.
- (v) The *point* or *eigenspectrum* of A is the set

$$\sigma_{\mathsf{p}}(A) = \big\{ \lambda \in \mathbb{C} \mid \ker(A - \lambda) \neq \{0\} \big\}.$$

(vi) An approximate eigenvalue of A is a complex number λ for which there exists a sequence of unit vectors $(v_n)_{n \in \mathbb{N}} \subset \mathcal{H}$ such that

$$\lim_{n \to \infty} (A - \lambda) v_n = 0.$$

The set $\sigma_{ap}(A)$ is the set of all approximate eigenvalues.

3.7.5 Evidently, $\sigma(A) = \mathbb{C} \setminus \varrho(A)$ and $\sigma_p(A) \subset \sigma_{ap}(A) \subset \sigma(A)$, and these may all be strict inclusions. Note that $A - \lambda$ is bounded for any $\lambda \in \mathbb{C}$, so the open mapping theorem **??** implies that $(A - \lambda)^{-1} \in \mathfrak{B}(\mathcal{H})$ when $\lambda \in \varrho(A)$. We call the map

$$R_{\bullet}(A) : \varrho(A) \to \mathfrak{B}(\mathcal{H}), \quad R_{\lambda}(A) = (A - \lambda)^{-1}$$

the *resolvent* of A, not to be confused with the resolvent set $\varrho(A)$. To keep the notation clean, we often briefly write R_{λ} for $R_{\lambda}(A)$ and leave implicit that R_{λ} depends on A.

First, we prove some topological properties of the spectrum and resolvent. Recall the following lemma, which generalizes the geometric series.

3.7.6 Lemma (Carl Neumann) Let $A \in \mathfrak{B}(\mathfrak{H})$. If ||A|| < 1, then I - A is invertible,

$$(I - A)^{-1} = \sum_{n=0}^{\infty} A^n,$$

and

$$\|(I-A)^{-1}\| \le \frac{1}{1-\|A\|}.$$

Proof. Since ||A|| < 1 and $||A^n|| \leq ||A||^n$ by submultiplicativity of the operator norm, we know $\sum_{n=0}^{\infty} ||A^n|| < \infty$. This implies that the family $(A^n)_{n \in \mathbb{N}}$ is absolutely summable, so $\sum_{n=0}^{\infty} A^n$ exists. Furthermore, for every $N \in \mathbb{N}$ we have

$$(I-A)\sum_{n=0}^{N}A^{n} = \left(\sum_{n=0}^{N}A^{n}\right)(I-A) = \sum_{n=0}^{N}A^{n} - \sum_{n=1}^{N+1}A^{n} = I - A^{N+1},$$

which implies that

$$\lim_{N \to \infty} (I - A) \sum_{n=0}^{N} A^n = \lim_{N \to \infty} \left(\sum_{n=0}^{N} A^n \right) (I - A) = I.$$

By continuity of multiplication in $\mathfrak{B}(\mathcal{H})$ one gets

$$(I-A)\sum_{n=0}^{\infty}A^n = \left(\sum_{n=0}^{\infty}A^n\right)(I-A) = I,$$

which proves that I - A is invertible and $(I - A)^{-1} = \sum_{n=0}^{\infty} A^n$.

Finally, one concludes by the triangle inequality and submultiplicativity of the operator norm

$$\|(I-A)^{-1}\| \leq \sum_{n=0}^{\infty} \|A^n\| \leq \sum_{n=0}^{\infty} \|A\|^n = \frac{1}{1-\|A\|}.$$

3.7.7 Proposition Let $A \in \mathfrak{B}(\mathcal{H})$.

(i) For any $\lambda \in \varrho(A)$, one has

$$B_{\|R_{\lambda}\|^{-1}}(\lambda) \subset \varrho(A)$$

Hence, $\varrho(A) \subset \mathbb{C}$ is open.

(ii) The spectrum $\sigma(A)$ is compact and

$$\sigma(A) \subset \bar{B}_{\|A\|}(0) \; .$$

(iii) If the complex number λ satisfies $|\lambda| > ||A||$, then $\lambda \in \varrho(A)$ and

$$R_{\lambda} = -\frac{1}{\lambda} - \sum_{n=1}^{\infty} \lambda^{-n-1} A^n ,$$

where convergence is with respect to the operator norm.

Proof. ad (i). Fix $\lambda \in \varrho(A)$ and set $r = ||R_{\lambda}||^{-1}$. Let $\mu \in B_r(\lambda)$. Then

$$\|(\mu - \lambda)R_{\lambda}\| = |\mu - \lambda| \|R_{\lambda}\| < 1$$

Thus, by Lemma 3.7.6, one knows that $I - (\mu - \lambda)R_{\lambda}$ is invertible. Since $A - \lambda$ is invertible, the composition

$$(A - \lambda) \left(I - (\mu - \lambda) R_{\lambda} \right) = A - \mu$$

is invertible, which proves that $\mu \in \varrho(A)$. Hence $\varrho(A)$ is open.

ad (ii). Since $\varrho(A)$ is open, the complement $\sigma(A) = \mathbb{C} \setminus \varrho(A)$ is closed. Furthermore, if $|\lambda| > ||A||$, then $||\lambda^{-1}A|| < 1$, so $I - \lambda^{-1}A$ and hence $A - \lambda$ are invertible by Lemma 3.7.6. This implies that $\lambda \in \varrho(A)$, so $\sigma(A) \subset \overline{B}_{||A||}(0)$. Since $\sigma(A)$ is closed and bounded, it is compact.

ad (iii). If $|\lambda| > ||A||$, then $I - \lambda^{-1}A$ is invertible by Lemma 3.7.6 and

$$(I - \lambda^{-1}A)^{-1} = \sum_{n=0}^{\infty} \lambda^{-n}A^n.$$

Since $-\lambda (A-\lambda)^{-1} = (I-\lambda^{-1}A)^{-1}$, one obtains

$$R_{\lambda} = -\frac{1}{\lambda} \sum_{n=0}^{\infty} \lambda^{-n} A^n = -\frac{1}{\lambda} - \sum_{n=1}^{\infty} \lambda^{-n-1} A^n,$$

as desired.

Next, we prove some algebraic properties of the resolvent. Hereby, [A, B] = AB - BA denotes the commutator of two operators, as usual.

3.7.8 Proposition Let $A, B \in \mathfrak{B}(\mathfrak{H})$. Then the following holds true.

(i) The resolvent commutes with the operator which means that

$$[A, R_{\lambda}(A)] = 0$$
 for all $\lambda \in \varrho(A)$.

(ii) The values of the resolvent commute with each other that is

$$[R_{\lambda}(A), R_{\mu}(A)] = 0$$
 for all $\lambda, \mu \in \varrho(A)$.

(iii) (First resolvent identity) For all $\lambda, \mu \in \varrho(A)$

$$R_{\lambda}(A) - R_{\mu}(A) = (\lambda - \mu)R_{\lambda}(A)R_{\mu}(A)$$

(iv) (Second resolvent identity) For all $\lambda \in \varrho(A) \cap \varrho(B)$

$$R_{\lambda}(A) - R_{\lambda}(B) = R_{\lambda}(A) (B - A) R_{\lambda}(B)$$

Proof. ad (i). Obviously $[A, A - \lambda] = 0$, so

$$0 = R_{\lambda} [A, A - \lambda] R_{\lambda} = R_{\lambda} A - A R_{\lambda},$$

as desired.

ad (iii). We compute

$$(R_{\lambda} - R_{\mu})(A - \mu)(A - \lambda) = (R_{\lambda}A - \mu R_{\lambda})(A - \lambda) - (A - \lambda)$$
$$= (A - \mu)R_{\lambda}(A - \lambda) - (A - \lambda)$$
$$= \lambda - \mu,$$

where we used part (i) to commute R_{λ} past A in the second step. Now multiplying both sides with $R_{\lambda}R_{\mu}$ from the right yields the desired equality.

ad (ii). For $\lambda = \mu$, one obviously has $[A_{\lambda}, A_{\mu}] = 0$. For $\lambda \neq \mu$, one concludes from (ii)

$$R_{\mu}R_{\lambda} = \frac{R_{\mu} - R_{\lambda}}{\mu - \lambda} = \frac{R_{\lambda} - R_{\mu}}{\lambda - \mu} = R_{\lambda}R_{\mu},$$

so $[R_{\lambda}, R_{\mu}] = 0$ for $\lambda \neq \mu$ as well.

ad (iv). The last equality follows by

$$R_{\lambda}(A) (B-A) R_{\lambda}(B) = R_{\lambda}(A) \left((B-\lambda) - (A-\lambda) \right) R_{\lambda}(B) = R_{\lambda}(A) - R_{\lambda}(B) . \square$$

The resolvent $R_{\bullet}(A)$ also has some nice analytic properties which we are going to prove next.

3.7.9 Proposition The resolvent $R_{\bullet}(A) : \varrho(A) \to \mathfrak{B}(\mathcal{H}), \lambda \mapsto R_{\lambda}$ is continuous and complex differentiable with derivative given by

$$R_{\bullet}(A)': \ \varrho(A) \to \mathfrak{B}(\mathcal{H}), \ \lambda \mapsto \lim_{\mu \to \lambda} \frac{R_{\mu} - R_{\lambda}}{\mu - \lambda} = R_{\lambda}^2$$

Proof. Fix $\lambda \in \varrho(A)$ and $\varepsilon > 0$. Let $0 < |\mu - \lambda| < \delta$, where

$$\delta = \min\left(rac{arepsilon}{2 \left\|R_{\lambda}
ight\|^{2}}, rac{1}{2 \left\|R_{\lambda}
ight\|}
ight) \;.$$

Note that $\mu \in \varrho(A)$ by Proposition 3.7.7. Moreover, $\|(\mu - \lambda)R_{\lambda}\| < 1$, so $I - (\mu - \lambda)R_{\lambda}$ is invertible with norm less than $(1 - \|(\mu - \lambda)R_{\lambda}\|)^{-1}$ by Lemma 3.7.6. Now observe that the first resolvent identity can be rearranged to

$$R_{\mu} = R_{\lambda} [I - (\mu - \lambda) R_{\lambda}]^{-1} .$$

Hence

$$\begin{split} \|R_{\mu} - R_{\lambda}\| &\leq |\mu - \lambda| \, \|R_{\mu}\| \, \|R_{\lambda}\| \\ &\leq |\mu - \lambda| \, \|R_{\lambda}\|^{2} \, \|(I - (\mu - \lambda)R_{\lambda})^{-1}\| \\ &\leq \frac{|\mu - \lambda| \, \|R_{\lambda}\|^{2}}{1 - \|(\mu - \lambda)R_{\lambda}\|} \\ &< \frac{\varepsilon/2}{1 - 1/2} = \varepsilon \; . \end{split}$$

This proves that $\lambda \mapsto R_{\lambda}$ is continuous.

As for complex differentiability, we simply use the first resolvent identity and continuity to conclude

$$\lim_{\mu \to \lambda} \frac{R_{\mu} - R_{\lambda}}{\mu - \lambda} = \lim_{\mu \to \lambda} R_{\mu} R_{\lambda} = R_{\lambda}^{2}.$$

3.7.10 Proposition Let $A \in \mathfrak{B}(\mathfrak{H})$. Then $\lambda R_{\lambda} \to -I$ as $|\lambda| \to \infty$. In particular, $R_{\lambda} \to 0$ as $|\lambda| \to \infty$.

Proof. Fix $\varepsilon > 0$. For $|\lambda| > ||A||$, we have by Proposition 3.7.7 (iii)

$$\lambda R_{\lambda} = -I - \sum_{n=1}^{\infty} \lambda^{-n} A^n.$$

Since

$$\left\|\sum_{n=1}^{\infty} \lambda^{-n} A^n\right\| \leqslant \frac{\|A\|}{|\lambda| - \|A\|},$$

one sees that $\lambda R_\lambda \to -I$ as $|\lambda| \to \infty.$ Similarly, for $|\lambda| > \|A\|$ one has

$$||R_{\lambda}|| \leq \frac{1}{|\lambda|} + \frac{1}{|\lambda|} \sum_{n=1}^{\infty} ||\lambda^{-n} A^{n}|| \leq \frac{1}{|\lambda|} + \frac{1}{|\lambda|} \frac{||A||}{|\lambda| - ||A||}$$

which shows that $R_{\lambda} \to 0$ as $|\lambda| \to \infty$.

3.7.11 Proposition For all $v, w \in \mathcal{H}$, the map

$$\langle R_{\bullet}(A)v, w \rangle : \varrho(A) \to \mathbb{C}, \ \lambda \mapsto \langle R_{\lambda}v, w \rangle$$

is holomorphic with derivative

$$\langle R_{\bullet}(A)v, w \rangle' : \varrho(A) \to \mathbb{C}, \ \lambda \mapsto \langle R_{\lambda}^2 v, w \rangle.$$

Proof. Given $\lambda \in \varrho(A)$, we compute

$$\lim_{\mu \to \lambda} \frac{\langle R_{\mu}v, w \rangle - \langle R_{\lambda}v, w \rangle}{\mu - \lambda} = \lim_{\mu \to \lambda} \frac{\langle (\mu - \lambda) R_{\mu} R_{\lambda}v, w \rangle}{\mu - \lambda} = \lim_{\mu \to \lambda} \langle R_{\mu} R_{\lambda}v, w \rangle = \langle R_{\lambda}^2 v, w \rangle,$$

where we have used the first resolvent identity in the first step and continuity of the inner product in the last. $\hfill \Box$

3.7.12 Proposition The spectrum of an operator $A \in \mathfrak{B}(\mathfrak{H})$ is nonempty.

Proof. Suppose $\sigma(A) = \emptyset$, hence $\varrho(A) = \mathbb{C}$. The map

$$\mathbb{C} \to \mathbb{C}, \ \lambda \mapsto \langle R_{\lambda} v, w \rangle$$

then is entire for every $v, w \in \mathcal{H}$. Furthermore, one has for $||v||, ||w|| \leq 1$

$$|\langle R_{\lambda}v, w \rangle| \leq ||R_{\lambda}|| ||v|| ||w|| \leq ||R_{\lambda}|| .$$

Since $\lambda \mapsto ||R_{\lambda}||$ is continuous and $||R_{\lambda}|| \to 0$ as $|\lambda| \to \infty$, one sees that $||R_{\lambda}||$ is bounded. Hence $\langle R_{\bullet}v, w \rangle$ is a bounded entire function, which by Liouville's theorem implies that it is zero for every pair $v, w \in \mathcal{H}$ with ||v|| = ||w|| = 1. This entails that $R_{\lambda} = 0$ for every $\lambda \in \mathbb{C}$, which is a contradiction to R_{λ} being invertible. Hence $\sigma(A) \neq \emptyset$.

A.3.8. Unbounded linear operators

3.8.1 In this section let V, W always denote Banach spaces over the field $\mathbb{K} = \mathbb{R}$ or $\mathbb{K} = \mathbb{C}$. The symbols $\mathcal{H}, \mathcal{H}_1, ...$ will always stand for Hilbert spaces over \mathbb{K} .

3.8.2 Definition By an *unbounded* \mathbb{K} -*linear operator* or shortly by an *unbounded operator* from V to W we understand a linear map $A : Dom(A) \to W$ defined on a \mathbb{K} -linear subspace $Dom(A) \subset V$. As usual, Dom(A) is called the *domain* of the operator A. The space of unbounded \mathbb{K} -linear operators from V to W will be denoted $\mathfrak{L}_{\mathbb{K}}(V, W)$ or just $\mathfrak{L}(V, W)$.

3.8.3 Remark In this work, the term "unbounded" is meant in the sense of "not necessarily bounded". Sometimes we just say *linear operator* or even only *operator* instead of "unbounded linear operator".

3.8.4 Observe that besides the domain Dom(A) of an unbounded operator $A \in \mathfrak{L}(V, W)$ the kernel

$$\operatorname{Ker}(A) = \left\{ v \in \mathcal{V} \mid Av = 0 \right\} \subset \mathcal{V} ,$$

the *image*

$$\operatorname{Im}(A) = \{ w \in W \mid \exists v \in \operatorname{Dom}(A) : w = Av \} \subset W ,$$

and the graph

$$Gr(A) = \{(v, w) \in Dom(A) \times W \mid w = Av\} \subset V \times W$$

of A are all linear subspaces. We will frequently make use of this.

3.8.5 Definition An unbounded operator $A \in \mathfrak{L}(V, W)$ is called *densely defined* if Dom(A) is dense in V, and *closed* if the graph Gr(A) is closed in $V \times W$. The operator $A \in \mathfrak{L}(V, W)$ is called *closable* if the closure Gr(A) is the graph of an unbounded operator from V to W.

An operator $A \in \mathfrak{L}(V, W)$ is called an *extension* of $B \in \mathfrak{L}(V, W)$ if $\operatorname{Gr}(B) \subset \operatorname{Gr}(A)$. One writes in this situation $B \subset A$.

A.4. C^* -Algebras

A.4.1. Infinite tensor products

4.1.1 Infinite tensor products of Hilbert spaces were introduced by von Neumann (1939). They were motivated by mathematical physics where one needs to describe quantum systems with infinitely many degrees of freedom, see e.g. Emch (2009); Bratteli & Robinson (1997). The original construction of infinite tensor products was generalized to von Neumann and C^* -algebras by Guichardet (1966), Blackadar (1977), and others. Meanwhile, the topic has been studied in quite some detail in the operator algebra literature, see e.g. Nakagami (1970a,b); Størmer (1971). A purely algebraic or better categorical approach allowing the construction of infinite tensor products of modules over a given commutative ring has been given in (Chevalley, 1956, Sec. III.10). The work Ng (2013) is also in that spirit. We will essentially follow Chevalley (1956) and construct the infinite tensor product as a module universal with respect to multilinear maps. First we present the main algebraic construction, then we explain some of the subtleties which distinguish infinite from finite tensor products, and finally we construct infinite Hilbert tensor products and infinite tensor products of C^* -algebras.

4.1.2 Let R be a commutative ring and $(M_i)_{i\in I}$ a possibly infinite family of R-modules. Consider $\prod_{i\in I} M_i$, the product of the family $(M_i)_{i\in I}$ within the category of R-modules. For each $j \in I$ let $\pi_j : \prod_{i\in I} M_i \to M_j$ denote the natural projection onto the *j*-th factor and $\iota_j : M_j \hookrightarrow \prod_{i\in I} M_i$ the uniquely determined natural embedding such that

$$\pi_j \circ \iota_i = egin{cases} \operatorname{id}_{M_i} & ext{for } i=j ext{ and } \ 0 & ext{else.} \end{cases}$$

Given an *R*-module *N* one then understands by a *multilinear map* from $\prod_{i \in I} M_i$ to *N* a map $f : \prod_{i \in I} M_i \to N$ such that for each $j \in I$ and $x \in \prod_{i \in I} M_i$ with $\pi_j(x) = 0$ the map $M_j \to N$, $m \mapsto f(\iota_j(m) + x)$ is linear. The set of multilinear maps from $\prod_{i \in I} M_i$ to *N* will be denoted by $\mathfrak{Mlin}(\prod_{i \in I} M_i, N)$. It carries a natural structure of an *R*-module given by pointwise addition of multilinear maps and pointwise action of a scalar on a multilinear map that is by

$$f + g = \left(\prod_{i \in I} M_i \ni x \mapsto f(x) + g(x) \in N\right) \quad \text{and} \quad rf = \left(\prod_{i \in I} M_i \ni x \mapsto rf(x) \in N\right)$$

for all $f, g \in \mathfrak{Mlin}(\prod_{i \in I} M_i, N)$ and $r \in R$. Since for $j \in I$ and $x \in \prod_{i \in I} M_i$ with $\pi_j(x) = 0$ the maps $M_j \to N$, $m \mapsto (f + g)(\iota_j(m) + x) = f(\iota_j(m) + x) + g(\iota_j(m) + x)$ and $M_j \to N$, $m \mapsto rf(\iota_j(m) + x)$ are linear by assumption on f and g, the maps f + g and rf are multilinear again, so $\mathfrak{Mlin}(\prod_{i \in I} M_i, N)$ is an R-module indeed with zero element the constant function mapping to $0 \in N$. **4.1.3 Remarks** Before proceeding further let us make several explanations concerning the notation used.

(a) The space of multilinear maps $\mathfrak{Min}(\prod_{i\in I} M_i, N)$ actually depends on the family $(M_i)_{i\in I}$ and the *R*-module *N*, so in principle one should write $\mathfrak{Min}((M_i)_{i\in I}, N)$ instead of $\mathfrak{Min}(\prod_{i\in I} M_i, N)$. Nevertheless we stick to the latter notation since it is closer to standard notation for linear maps and since it will not lead to any confusion.

(b) In case the index set I has just two elements i_1, i_2 , one calls a multilinear map $\prod_{i \in I} M_i = M_{i_1} \times M_{i_2} \to N$ a bilinear map. If the cardinality of I is 3, one sometimes calls a multilinear map $\prod_{i \in I} M_i \to N$ a trilinear map.

(c) In the following, when saying that $(I_a)_{a \in A}$ is a partition of the set I we mean that each I_a is a non-empty subset of I, that $I_a \cap I_b = \emptyset$ for $a \neq b$ and that $\bigcup_{a \in A} I_a = I$. The empty family is regarded as a partition of the empty set.

(d) We will frequently use in this section the same symbol for maps with the same "universal" properties despite those maps might be strictly speaking different. For example, π_k will stand for the canonical projections $\prod_{i \in I} M_i \to M_k$ and $\prod_{j \in J} M_j \to M_k$ whenever $k \in J \subset I$. Likewise we use the same notation for the two canonical embeddings $M_k \hookrightarrow \prod_{i \in I} M_i$ and $M_k \hookrightarrow \prod_{j \in J} M_j$ defined in 4.1.2 and denote them both by ι_k .

4.1.4 Lemma (cf. (Chevalley, 1956, Sec. III.10, Lemma 1 & 2)) Assume that $(M_i)_{i \in I}$ is a family of *R*-modules, *N* an *R*-module, and $f : \prod_{i \in I} M_i \to N$ a mutilinear map.

- (i) If $g: N \to N'$ is an *R*-module map, then $g \circ f: \prod_{i \in I} M_i \to N'$ is multilinear.
- (ii) Let $J \subset I$ be non-empty, $y = (y_i)_{i \in I \setminus J}$ an element of the product $\prod_{i \in I \setminus J} M_i$, and $\iota_{J,y} : \prod_{j \in J} M_j \to \prod_{i \in I} M_i$ the unique map such that for all $x = (x_j)_{j \in J} \in (M_j)_{j \in J}$ and $k \in I$

$$\pi_{k} \circ \iota_{J,y}\left(x\right) = \begin{cases} x_{k} & \text{for } k \in J, \\ y_{k} & \text{for } k \in I \backslash J. \end{cases}$$

Then the composition $f \circ \iota_{J,x} : \prod_{j \in J} M_j \to N$ is multilinear.

(iii) Let $(I_a)_{a \in A}$ be a partition of the index set I which is assumed to be non-empty. Let $(N_a)_{a \in A}$ be a family of R-modules, $(g_a)_{a \in A}$ a family of multilinear maps $g_a : \prod_{i \in I_a} M_i \to N_a$, and $h : \prod_{a \in A} N_a \to N$ multilinear. Define $g : \prod_{i \in I} M_i \to \prod_{a \in A} N_a$ as the unique map such that

$$\pi_b \circ g = g_b \circ \pi_{I_b}$$
 for $b \in A_{I_b}$

where π_J for $J \subset I$ as on the right side stands for the projection $\pi_J : \prod_{i \in I} M_i \to \prod_{j \in J} M_j$ uniquely determined by $\pi_j \circ \pi_J = \pi_j$ for all $j \in J$. Then the composition $h \circ g : \prod_{i \in I} M_i \to N$ is multilinear.

Proof. ad (i). Let $j \in I$ and $x \in \prod_{i \in I} M_i$ with $\pi_j(x) = 0$. By multilinearity of f and linearity of g, the map $M_j \to N'$, $m \mapsto gf(\iota_j(m) + x)$ then has to be linear, hence $g \circ f$ is multilinear.

ad (ii). Let $j \in J$ and $x \in \prod_{i \in J} M_i$ with $\pi_j(x) = 0$. Then $\pi_j(\iota_{J,y}(x)) = 0$ and $f\iota_{J,y}(\iota_j(m) + x) = f(\iota_j(m) + \iota_{J,y}(x))$ for all $m \in M_j$ by construction of $\iota_{J,y}$. Hence the map $M_j \to N$, $m \mapsto f\iota_{J,y}(\iota_j(m) + x)$ is linear by multilinearity of f. This proves that $f \circ \iota_{J,y}$ is multilinear.

ad (iii). Given $j \in I$ let b be the unique element of A such that $j \in I_b$. Assume that $x \in \prod_{i \in I} M_i$ with $\pi_j(x) = 0$. By construction one has $\pi_j(\pi_{I_b}(x)) = 0$. Now let $y \in \prod_{a \in A} N_a$ such that

$$\pi_a(y) = \begin{cases} 0 & \text{for } a = b, \\ g_a \pi_{I_a}(x) & \text{for } a \neq b. \end{cases}$$

One then obtains for $m \in M_i$

$$\pi_a g(\iota_j(m) + x) = \begin{cases} g_b \pi_{I_b}(\iota_j(m) + x) = g_b(\iota_j(m) + \pi_{I_b}(x)) & \text{for } a = b, \\ g_a \pi_{I_a}(x) = \pi_a(y) & \text{for } a \neq b. \end{cases}$$

Hence

$$hg(\iota_j(m) + x) = h\big(\iota_b\big(g_b(\iota_j(m) + \pi_{I_b}(x)\big) + y\big) ,$$

and the map $M_j \to N$, $m \mapsto hg(\iota_j(m) + x)$ is linear as the composition of two linear maps. \Box

4.1.5 Lemma Assume to be given a non-empty family of R-modules $(M_i)_{i \in I}$ and a partition $(I_a)_{a \in A}$ of the index set I. Then there exists a natural ismorphism

$$\kappa_{I,A} : \prod_{i \in I} M_i \to \prod_{a \in A} \prod_{i \in I_a} M_i$$

uniquely determined by the condition that $\pi_a \circ \kappa_{I,A} = \pi_{I_a}$ for all $a \in A$.

Proof. By the universal property of the product the *R*-module map $\kappa = \kappa_{I,A} : \prod_{i \in I} M_i \to \prod_{a \in A} \prod_{i \in I_a} M_i$ exists and is uniquely determined by the requirement that $\pi_a \circ \kappa_{I,A} = \pi_{I_a}$ for all $a \in A$. Naturality also follows from the universal property of the product. It remains to show that κ is an isomorphism. By construction, $\pi_i(x) = \pi_i \pi_a \kappa(x) = 0$ for all $i \in I$ and $a(i) \in A$ such that $i \in I_{a(i)}$, hence x = 0. So κ is injective. It is also surjective. To see this pick $x_a \in \prod_{i \in I_a} M_i$ for each $a \in A$. With a(i) for $i \in I$ defined as before put $x = (\pi_i(x_{a(i)}))_{i \in I}$. Then, by construction, $\pi_i \pi_a \kappa(x) = \pi_i \pi_a(x) = \pi_i(x)$ for all $a \in A$ and $i \in I_a$, hence $(\pi_a \kappa(x))_{a \in A} = (x_a)_{a \in A}$ and κ is surjective.

4.1.6 Proposition (Exponential law for multilinear maps) Let $(M_i)_{i \in I}$ be a family of *R*-modules over a commutative ring *R*, *N* an *R*-module, and assume that $J \subset I$ is a non-empty subset such that the complement $K = I \setminus J$ is also non-empty. Then the map

$$\eta_{I,J}:\mathfrak{M}\mathfrak{lin}\left(\prod_{j\in J}M_j,\mathfrak{M}\mathfrak{lin}\left(\prod_{k\in K}M_k,N\right)\right)\to\mathfrak{M}\mathfrak{lin}\left(\prod_{i\in I}M_i,N\right),$$
$$f\mapsto\left(\prod_{i\in I}M_i\ni(x_i)_{i\in I}\mapsto f\left((x_j)_{j\in J}\right)\left((x_k)_{k\in K}\right)\in N\right)$$

is an isomorphism which is natural in $(M_i)_{i \in I}$ and N.

Proof. We first show that $\eta = \eta_{I,J}$ is linear. To this end let $f, g \in \mathfrak{Mlin}\left(\prod_{j \in J} M_j, \mathfrak{Mlin}\left(\prod_{k \in K} M_k, N\right)\right)$ and $r \in R$. Then, for all $x = (x_i)_{i \in I} \in \prod_{i \in I} M_i$,

$$(\eta(f+g))(x) = (f+g)((x_j)_{j\in J})((x_k)_{k\in K}) = (f((x_j)_{j\in J}) + g((x_j)_{j\in J}))((x_k)_{k\in K}) = = f((x_j)_{j\in J})((x_k)_{k\in K}) + g((x_j)_{j\in J})((x_k)_{k\in K}) = (\eta f)(x) + (\eta g)(x) = (\eta f + \eta g)(x)$$

and

$$(\eta(rf))(x) = (rf)((x_j)_{j \in J})((x_k)_{k \in K}) = (rf((x_j)_{j \in J}))((x_k)_{k \in K}) = r(f((x_j)_{j \in J})((x_k)_{k \in K})) = r(\eta f(x)) = (r(\eta f))(x) .$$

Hence η is an *R*-module map.

Next we show that η is an isomorphism by constructing an inverse. Given $f \in \mathfrak{Mlin}(\prod_{i \in I} M_i, N)$ we define $f^{\sharp} : \mathfrak{Mlin}(\prod_{j \in J} M_j) \to \mathfrak{Mlin}(\prod_{k \in K} M_k, N)$ by the requirement that

$$f^{\sharp}(y)(z) = f(x_{y,z}) \quad \text{for all } y = (y_j)_{j \in J} \text{ and } z = (z_k)_{k \in K} \;,$$

where $x_{y,z}$ is the element of $\prod_{i \in I} M_i$ uniquely determined by

$$\pi_i(x_{y,z}) = \begin{cases} y_i & \text{for } i \in J, \\ z_i & \text{for } i \in K. \end{cases}$$

One thus obtains an R-module map

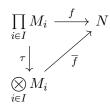
$$(-)_{I,J}^{\sharp}:\mathfrak{Mlin}\left(\prod_{i\in I}M_{i},N\right)\to\mathfrak{Mlin}\left(\prod_{j\in J}M_{j},\mathfrak{Mlin}\left(\prod_{k\in K}M_{k},N\right)\right),\quad f\mapsto f^{\sharp}$$

which by construction is inverse to $\eta_{I,J}$.

Naturality of $\eta_{I,J}$ in $(M_i)_{i \in J}$ and N is clear by definition.

4.1.7 Definition Let $(M_i)_{i \in I}$ be a family of R-modules over a commutative ring R. By a *tensor* product of $(M_i)_{i \in I}$ one understands an R-module $\bigotimes_{i \in I} M_i$ together with a multilinear map τ : $\prod_{i \in I} M_i \to \bigotimes_{i \in I} M_i$ such that the following universal property is fulfilled:

(ITensor) For every *R*-module *N* and every multilinear map $f : \prod_{i \in I} M_i \to N$ there exists a unique *R*-module map $\overline{f} : \bigotimes_{i \in I} M_i \to N$ such that the diagram



commutes.

The linear map \overline{f} making the diagram comute will sometimes be called the *linearization* of the multilinear map f.

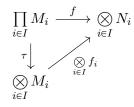
Given a tensor product $(\bigotimes_{i \in I} M_i, \tau)$, we will usually denote the image of an element $(x_i)_{i \in I} \in \prod_{i \in I} M_i$ under the map τ by $\bigotimes_{i \in I} x_i$.

4.1.8 Remarks (a) Strictly speaking, a tensor product of a family $(M_i)_{i \in I}$ of R-modules is a pair $(\bigotimes_{i \in I} M_i, \tau)$ having the above properties. By slight abuse of language, one usually denotes a tensor product just by its first component, the R-module $\bigotimes_{i \in I} M_i$. When helpful for clarity, the associated map $\tau : \prod_{i \in I} M_i \to \bigotimes_{i \in I} M_i$ will be denoted by $\tau_{(M_i)_{i \in I}}$ or by τ_I .

(b) In the case where the index set I of the family $(M_i)_{i \in I}$ is infinite, one sometimes calls $\bigotimes_{i \in I} M_i$ an *infinite tensor product*.

4.1.9 Theorem Let $(M_i)_{i \in I}$ be a family of *R*-modules over a commutative ring *R*. Then the following holds true.

- (i) A tensor product $\bigotimes_{i \in I} M_i$ of the family $(M_i)_{i \in I}$ exists and is unique up to isomorphism. If I is the empty set, then $\bigotimes_{i \in I} M_i = R$, if I contains a single element i_{\circ} , then $\bigotimes_{i \in I} M_i = M_{i_{\circ}}$.
- (ii) If $(N_i)_{i \in I}$ is a second family of *R*-modules and $(f_i)_{i \in I}$ a family *R*-module maps $f_i : M_i \to N_i$, then there exists a unique linear map $\bigotimes_{i \in I} f_i : \bigotimes_{i \in I} M_i \to \bigotimes_{i \in I} N_i$ making the diagram



commute, where $f: \prod_{i \in I} M_i \to \bigotimes_{i \in I} N_i$ is the multilinear map $(x_i)_{i \in I} \mapsto \bigotimes_{i \in I} f_i(x_i)$.

(iii) Let J ⊂ I be a finite non-empty subset set such that M_j is isomorphic to R for all j ∈ J. Denote for each j ∈ J by 1_j the image of the unit 1 ∈ R under the isomorphism R ≅ M_j and by 1_J the family (1_j)_{j∈J}. Moreover, for every family y = (y_j)_{j∈J} let ι_{J,y} : ∏_{i∈I\J} M_i → ∏_{i∈I} M_i be the map which associates to x ∈ ∏_{i∈I\J} M_i the family (x_i)_{i∈I} such that x_i = π_i(x) for i ∈ I\J and x_i = y_i for i ∈ J. Then the linearization ī_{J,1J} : ⊗_{i∈I\J} M_i → ⊗_{i∈I} M_i of the multilinear map τ_I ∘ ι_{J,1J} : ∏_{i∈I\J} M_i → ⊗_{i∈I} M_i is an isomorphism.

Proof. ad (i). By its universal property, the tensor product of the family $(M_i)_{i \in I}$ is uniquely determined up to isomorphism. Hence it remains to show the existence of the tensor product. To this end consider the free R-module over the set $\prod_{i \in I} M_i$ and denote it by F. Let $\delta : \prod_{i \in I} M_i \hookrightarrow F$ be the canonical injection and U be the submodule of F spanned by the elements

$$\delta\big(\iota_j(ry_j+z_j)+(x_i)_{i\in I}\big)-r\delta\big(\iota_j(y_j)+(x_i)_{i\in I}\big)-\delta\big(\iota_j(z_j)+(x_i)_{i\in I}\big),$$

where $j \in I$, $y_j, z_j \in M_j$, $r \in R$, and $(x_i)_{i \in I} \in \pi_j^{-1}(0)$. Then put $\bigotimes_{i \in I} M_i = F/U$ and define τ as the composition of the canonical projection $\pi : F \to \bigotimes_{i \in I} M_i$ with $\delta : \prod_{i \in I} M_i \to F$. By construction, τ is multilinear. Assume that N is an R-module and $f : \prod_{i \in I} M_i \to N$ is a multilinear map. By the universal property of free R-modules, f lifts to a unique R-linear map $f' : F \to N$ such that $f = f' \circ \delta$. By multilinearity of f, the map f' vanishes on the submodule U, hence descends to an R-linear $\overline{f} : \bigotimes_{i \in I} M_i \to N$ such that $f' = \overline{f} \circ \pi$. Hence $f = f' \circ \delta = \overline{f} \circ \pi \circ \delta = \overline{f} \circ \tau$. By surjectivity of δ and uniqueness of f', \overline{f} is the unique R-linear map satisfying $f = \overline{f} \circ \tau$. Hence $(\bigotimes_{i \in I} M_i, \tau)$ is a tensor product of the family $(M_i)_{i \in I}$.

In case $I = \emptyset$, the cartesian product $\prod_{i \in I} M_i$ is final in the category of sets, hence consists of only one element \star only. This means in particular that for an R-module N any map $f : \prod_{i \in I} M_i = \{\star\} \to N$ is multilinear. Put $\bigotimes_{i \in I} M_i = R$ and let $\tau : \{\star\} \to R$ be the map $\star \mapsto 1$. Now let $\overline{f} : R \to N$ be the unique linear map such that $\overline{f}(1) = f(\star)$. Then $f = \overline{f} \circ \tau$ and the pair (R, τ) fulfills the universal property of the tensor product.

If I is a singleton with unique element i_0 , then $\prod_{i \in I} M_i = M_{i_0}$ and a map $f : \prod_{i \in I} M_i \to N$ is multilinear if and only if f as a map from M_{i_0} to N is linear. This implies that the pair $(M_{i_0}, \operatorname{id}_{M_{i_0}})$ then is a tensor product for the family $(M_i)_{i \in I}$.

ad (ii). This is an immediate consequence of the universal property of the tensor product.

ad (iii). We construct an inverse to $\bar{\iota}_{J,1_J} : \bigotimes_{i \in I \setminus J} M_i \to \bigotimes_{i \in I} M_i$. Let $x = (x_i)_{i \in I}$ be an element of $\prod_{i \in I} M_i$ and put

$$\lambda(x) = \left(\prod_{j \in J} x_j\right) \cdot \bigotimes_{i \in I \setminus J} x_i \left(\prod_{j \in J} x_j\right) \cdot \tau_{I \setminus J}((x_i)_{i \in I \setminus J})$$

Then $\lambda : \prod_{i \in I} M_i \to \bigotimes_{i \in \backslash J} M_i$ is multilinear by construction, hence factors through a linear map $\overline{\lambda} : \bigotimes_{i \in I} M_i \to \bigotimes_{i \in I \setminus J} M_i$. By definition, $\overline{\lambda}$ is a left inverse of $\overline{\iota}_{J,1_J}$. It is also a right inverse since for all $(x_i)_{i \in I} \in \prod_{i \in I} M_i$ by multilinearity of τ_I

$$\overline{\iota}_{J,1_J} \circ \overline{\lambda} \circ \tau_I \left((x_i)_{i \in I} \right) = \overline{\iota}_{J,1_J} \left(\left(\prod_{j \in J} x_j \right) \cdot \bigotimes_{i \in I \setminus J} x_i \right) = \left(\prod_{j \in J} x_j \right) \cdot \left(\overline{\iota}_{J,1_J} \circ \tau_{I \setminus J} \left((x_i)_{i \in I \setminus J} \right) \right) = \\ = \left(\prod_{j \in J} x_j \right) \cdot \left(\tau_I \circ \iota_{J,1_J} \left((x_i)_{i \in I \setminus J} \right) \right) = \tau_I \circ \iota_{J,(x_j)_{j \in J}} \left((x_i)_{i \in I \setminus J} \right) = \tau_I \left((x_i)_{i \in I} \right)$$

and since by construction of the tensor product the image of τ_I is a generating system for the R-module $\bigotimes_{i \in I} M_i$.

4.1.10 Lemma Assume that $(M_i)_{i \in I}$ is a finite family of R-modules such that for every $i \in I$ a generating set S_i of the R-module M_i has been given. Then the set $S = \tau (\prod_{i \in I} S_i)$ is a generating set of the tensor product $\bigotimes_{i \in I} M_i$.

Proof. By construction of the tensor product in the proof of Theorem 4.1.9 it is clear that a generating set of $\bigotimes_{i \in I} M_i$ is given by the set of elements of the form $\bigotimes_{i \in I} x_i$ where $(x_i)_{i \in I} \in \prod_{i \in I} M_i$. Each of the x_i can now be represented in the form

$$x_i = \sum_{k=1}^{n_i} r_{i,k} s_{i,k}$$
 with $r_{i,1}, \dots, r_{i,n_i} \in R, s_{i,1}, \dots, s_{i,n_i} \in S_i$.

Hence, by multilinearity of τ and with $I = \{i_1, \ldots, i_d\}$,

$$\otimes_{i \in I} x_i = \tau \left((x_i)_{i \in I} \right) = \sum_{k_{i_1}=1}^{n_{i_1}} \cdots \sum_{k_{i_d}=1}^{n_{i_d}} r_{i_1, k_{i_1}} \cdot \ldots \cdot r_{i_d, k_{i_d}} \cdot \tau \left((s_{i, k_i})_{i \in I} \right) ,$$

so $\bigotimes_{i \in I} x_i$ is a linear combination of elements of S and the claim is proved.

4.1.11 Lemma Let $(M_i)_{i \in I}$ be a family of R-modules, $(I_a)_{a \in A}$ a finite partition of the index set I, and N an R-module. For $a \in A$ put $N_a = \bigotimes_{i \in I_a} M_i$ and let $\tau_a : \prod_{i \in I_a} M_i \to N_a$ denote the canonical map. Assume that $f : \prod_{a \in A} \prod_{i \in I_a} M_i \to N$ is a map which is componentwise multilinear in the following sense.

(CM) Let $b \in A$ and $y = (y_a)_{a \in A} \in \prod_{a \in A} \prod_{i \in I_a} M_i$ a family with $y_b = 0$. If for all $j \in I_b$ and families $x = (x_i)_{i \in I_b} \in \prod_{i \in I_b} M_i$ with $x_j = 0$ the map

$$M_j \to N, \ m \mapsto f(\iota_b(\iota_j(m) + x) + y)$$

is linear, then f factors through $(\tau_a)_{a \in A} : \prod_{a \in A} \prod_{i \in I_a} M_i \to \prod_{a \in A} N_a$. More precisely, there exists a unique multilinear map $\overline{f} : \prod_{a \in A} N_a \to N$ such that

$$f = f \circ (\tau_a)_{a \in A} .$$

Proof. We prove the claim by induction on the cardinality of A. If A is a singleton, then $\prod_{a \in A} \prod_{i \in I_a} M_i$ canonically coincides with $\prod_{i \in I} M_i$ and $f : \prod_{i \in I_a} M_i \to N$ is multilinear, hence by the universal property of the tensor product there exists a unique linear map $\overline{f} : N_a \to N$ such that $f = \overline{f} \circ \tau_a$.

Now assume that the claim holds whenever the cardinality of the index set A is $\leq n$ for some $n \in \mathbb{N}^*$. Assume to be given initial data $(M_i)_{i \in I}$ and N, a partition $(I_a)_{a \in A}$ of A with |A| = n + 1 and componentwise multilinear map $f : \prod_{a \in A} \prod_{i \in I_a} M_i \to N$. Fix $a \in A$ and put $B = A \setminus \{a\}$. Let $x = (x_i)_{i \in I_a} M_i$ and \widetilde{x} be the element of $\prod_{d \in A} \prod_{i \in I_d} M_i$ such that

$$\pi_d(\widetilde{x}) = \begin{cases} x & \text{for } d = a \\ 0 & \text{else }. \end{cases}$$

The map

$$f_x : \prod_{b \in B} \prod_{i \in I_b} M_i \to N, \ y \mapsto f(\iota_B(y) + \widetilde{x})$$

then is componentwise multilinear. Hence by inductive assumption there exists a unique multilinear map $\overline{f_x} : \prod_{b \in B} N_b \to N$ such that $f_x = \overline{f_x} \circ (\tau_b)_{b \in B}$. By assumption on f the map $\prod_{i \in I_a} M_i \to \mathfrak{Map}\left(\prod_{b \in B} \prod_{i \in I_b} M_i, N\right)$, $x \mapsto f_x$ is multilinear which implies multilinearity of

$$\overline{f_{\bullet}}:\prod_{i\in I_a}M_i\to\mathfrak{M}\mathfrak{lin}\left(\prod_{b\in B}N_b,N\right), \ x\mapsto\overline{f_x}$$

Let $F : N_a \to \mathfrak{M}\mathfrak{lin}(\prod_{b \in B} N_b, N)$ be its linearization. Application of the exponential law for multilinear maps, Proposition 4.1.6, now gives a multilinear map $\eta(F) : \prod_{d \in A} N_d \to N$ which we denote by \overline{f} . Given a family $(x_d)_{d \in A}$ of families $x_d = (x_i)_{i \in I_d}$ one checks

$$\overline{f}\left(\left(\tau_d(x_d)\right)_{d\in A}\right) = F\left(\tau_a(x_a)\right)\left(\left(\tau_b(x_b)\right)_{b\in B}\right) = \overline{f}_{x_a}\left(\left(\tau_b(x_b)\right)_{b\in B}\right) = f_{x_a}\left((x_b)_{b\in B}\right) = f\left((x_d)_{d\in A}\right) \ .$$

Hence $\overline{f} \circ (\tau_d)_{d \in A} = f$. To finish the induction step it remains to prove uniqueness. So let \overline{g} : $\prod_{d \in A} N_d \to N$ be another multilinear map such that $\overline{g} \circ (\tau_d)_{d \in A} = f$ and consider the induced linear map $\overline{g}^{\sharp} = \eta^{-1}(\overline{g}) : N_a \mapsto \mathfrak{Mlin}(\prod_{b \in B} N_b, N)$. Then for every $x \in \prod_{i \in I_a} M_i$ the relation

$$\overline{g}^{\sharp}(\tau_a(x)) \circ (\tau_b)_{b \in B} = f_x = f_x \circ (\tau_b)_{b \in B}$$

is satisfied. Hence $\overline{g}^{\sharp}(\tau(x)) = \overline{f}_x$ for all $x \in \prod_{i \in I_a} M_i$ which entails that \overline{g}^{\sharp} coincides with F. By Proposition 4.1.6 one obtains $\overline{g} = \overline{f}$. This finishes the induction step and the lemma is proved. \Box

4.1.12 Proposition Let $(M_i)_{i \in I}$ be a family of *R*-modules and $(I_a)_{a \in A}$ a finite partition of the index set *I*. Then there exists a natural isomorphism

$$\alpha_{I,A}: \bigotimes_{i\in I} M_i \to \bigotimes_{a\in A} \bigotimes_{i\in I_a} M_i.$$

Proof. Put $N_a = \bigotimes_{i \in I_a} M_i$ for $a \in A$ and let $\tau_a : \prod_{i \in I_a} M_i \to N_a$ be the canonical map to the tensor product. Let $\tau_A : \prod_{a \in A} N_a \to \bigotimes_{a \in A} N_a$ be the canonical map to the tensor product of the modules N_a . Define $\tau_{I,A} : \prod_{i \in I} M_i \to \prod_{a \in A} N_a$ as the unique map so that $\pi_a \circ \tau_{I,A} = \tau_a \circ \pi_{I_a}$ for all $a \in A$. By construction $\tau_{I,A} = (\tau_a)_{a \in A} \circ \kappa_{I,A}$, where $\kappa_{I,A} : \prod_{i \in I} M_i \to \prod_{a \in A} M_i$ is the natural isomorphism from Lemma 4.1.5. The composition $\tau_A \circ \tau_{I,A}$ then is multilinear by Lemma 4.1.4 (iii), hence factors through a linear map $\alpha_{I,A} : \bigotimes_{i \in I} M_i \to \bigotimes_{a \in A} N_a$ that is

$$\tau_A \circ (\tau_a)_{a \in A} \circ \kappa_{I,A} = \alpha_{I,A} \circ \tau_I . \tag{A.4.1.1}$$

Naturality of $\alpha_{I,A}$ in $(M_i)_{i\in I}$ is clear by definition so it remains to construct an inverse to $\alpha_{I,A}$. Consider the composition $\tau_I \circ \kappa^{-1} : \prod_{a \in A} \prod_{i \in I_a} M_i \to \bigotimes_{i \in I} M_i$. Assume that $a \in A$ and $(y_b)_{b \in A \setminus \{a\}} \in \prod_{b \in A \setminus \{a\}} \prod_{i \in I_b} M_i$ have been chosen. Let $y_a \in \prod_{i \in I_a} M_i$ be 0, put $\widetilde{y} = (y_d)_{d \in A} \in \prod_{d \in A} \prod_{i \in I_d} M_i$, and let $y \in \prod_{i \in I} M_i$ be the family such that $\pi_i(y) = \pi_i(y_{a(i)})$ for all $i \in I$, where a(i) denotes the unique element of A such that $i \in I_{a(i)}$. In other words let $y = \kappa^{-1}(\widetilde{y})$. For every $j \in I_a$ and $x = (x_i)_{i \in I_a} M_i$ with $\pi_j(x) = 0$ the map

$$M_j \to \bigotimes_{i \in I} M_i, \ m \mapsto \tau_I \circ \kappa^{-1} \left(\iota_a(\iota_j(m) + x) + \widetilde{y} \right) = \tau_I \left(\iota_j(m) + \iota_{I_a}(x) + y \right)$$

then is multilinear since τ_I is multilinear and $\pi_j(\iota_{I_a}(x) + y) = \pi_j(x) + \pi_j(y_a) = 0$. Hence $\tau_I \circ \kappa^{-1}$ is componentwise multilinear and therefore, by Lemma 4.1.11, factors through the map $(\tau_a)_{a \in A} : \prod_{a \in A} \prod_{i \in I_a} M_i \to \prod_{a \in A} N_a$ which means that

$$\tau_I \circ \kappa^{-1} = \lambda_{I,A} \circ (\tau_a)_{a \in A} \tag{A.4.1.2}$$

for some uniquely defined multilinear map $\lambda_{I,A}: \prod_{a\in A} N_a \to \bigotimes_{i\in I} M_i$. Let

$$\overline{\lambda}_{I,A} : \bigotimes_{a \in A} N_a \to \bigotimes_{i \in I} M_i$$

be the linearization of $\lambda_{I,A}$. We claim that $\overline{\lambda_{I,A}}$ is inverse to $\alpha_{I,A}$. By definition of $\overline{\lambda_{I,A}}$ and Eqs. (A.4.1.1) and (A.4.1.2) one concludes

$$\overline{\lambda_{I,A}} \circ \alpha_{I,A} \circ \tau_I = \overline{\lambda_{I,A}} \circ \tau_A \circ (\tau_a)_{a \in A} \circ \kappa_{I,A} = \lambda_{I,A} \circ (\tau_a)_{a \in A} \circ \kappa_{I,A} = \tau_I .$$

Since the image of τ_I generates $\bigotimes_{i \in I} M_i$ as an *R*-module, $\overline{\lambda_{I,A}}$ has to be left inverse to $\alpha_{I,A}$. Using Eqs. (A.4.1.1) and (A.4.1.2) again compute

$$\alpha_{I,A} \circ \overline{\lambda_{I,A}} \circ \tau_A \circ (\tau_a)_{a \in A} = \alpha_{I,A} \circ \lambda_{I,A} \circ (\tau_a)_{a \in A} = \alpha_{I,A} \circ \tau_A \circ \kappa_{I,A}^{-1} = \tau_A \circ (\tau_a)_{a \in A} .$$

Since by Lemma 4.1.10 the image of $\tau_A \circ (\tau_a)_{a \in A}$ generates $\bigotimes_{a \in A} \bigotimes_{i \in I_a} M_i$, the equality

$$\alpha_{I,A} \circ \overline{\lambda_{I,A}} = \mathrm{id}_{\bigotimes_{a \in A} \bigotimes_{i \in I_a} M_i}$$

follows and the proposition is proved.

4.1.13 Proposition and Definition Let $(A_i)_{i \in I}$ be a family of *R*-algebras. Then the tensor product $A = \bigotimes_{i \in I} A_i$ carries in a natural way the structure of an *R*-algebra where the product map is defined by

:
$$A \times A \to A$$
, $(\bigotimes_{i \in I} a_i, \bigotimes_{i \in I} b_i) \mapsto \bigotimes_{i \in I} (a_i \cdot b_i)$.

In case each of the algebras A_i is commutative, then A is commutative as well. Likewise, if each A_i is unital and 1_i denotes the unit element of A_i , then A is unital with unit given by $1 = \bigotimes_{i \in I} 1_i$. One calls A the tensor product algebra of the family of algebras $(A_i)_{i \in I}$.

Proof. The map

$$\prod_{(i,k)\in I\times\{1,2\}} A_i \to A, \ (a_{i,k})_{(i,k)\in I\times\{1,2\}} \mapsto \bigotimes_{i\in I} (a_{i,1}\cdot a_{i,2})$$

is multilinear by bilinearity of the product maps on the A_i and multilinearity of τ_I , so factors through a linear map $\mu : A \otimes A \cong \bigotimes_{(i,k) \in I \times \{1,2\}} A_i \to A$. Composition of μ with the canonical bilinear map $A \times A \to A \otimes A$ gives the product map $\cdot : A \times A \to A$ and shows that the product on A is well-defined. By construction, the product map \cdot is bilinear. Given $\bigotimes_{i \in I} a_i, \bigotimes_{i \in I} b_i, \bigotimes_{i \in I} c_i \in A$ one computes

$$\left(\bigotimes_{i\in I}a_i\cdot\bigotimes_{i\in I}b_i\right)\cdot\bigotimes_{i\in I}c_i=\bigotimes_{i\in I}((a_i\cdot b_i)\cdot c_i)=\bigotimes_{i\in I}(a_i\cdot (b_i\cdot c_i))=\bigotimes_{i\in I}a_i\cdot\left(\bigotimes_{i\in I}b_i\cdot\bigotimes_{i\in I}c_i\right).$$

This entails that the product on A is associative. In the same way one shows that A is commutive respectively unital if each of the A_i is.

4.1.14 As we have seen, the infinite tensor product construction works well for objects of algebraic categories like R-modules, vector spaces or R-algebras. As soon as a topologies compatible with the algebraic structure come in it becomes difficult and sometimes even impossible to construct or even define

A.5. Manifolds

A.5.1. Pro-manifolds

A.5.2. Hilbert manifolds

5.2.1 In this section we will describe several examples of Hilbert manifolds.

5.2.2 Example Let \mathcal{H} be a Hilbert space over the field \mathbb{K} of real or complex numbers and $\omega : \mathcal{H} \to \mathbb{R}$ a continuous nonzero real linear form on \mathcal{H} . Then the *sphere*

$$\mathbb{S}(\mathcal{H}) = \left\{ v \in \mathcal{H} \mid \|v\| = 1 \right\}$$

is a real analytic Hilbert manifold modelled on the real Hilbert space $\ker \omega$. The sphere has tangent bundle

$$T\mathbb{S}(\mathcal{H}) = \left\{ (v, w) \in \mathbb{S}(\mathcal{H}) \times \mathcal{H} \mid \mathfrak{Re}\langle v, w \rangle = 0 \right\} \,.$$

A.5.3. The Graßmann manifold of a Banach space

5.3.1 Throughout this section we denote by E a Banach space over the field $\mathbb{K} = \mathbb{R}$ or $= \mathbb{C}$. The main object of study of this section then is the space $\mathbb{G}E$ of closed \mathbb{K} -linear subspaces of E. It is called the *Graßmann manifold* or *Graßmannian* of E. Let us equip $\mathbb{G}E$ with a natural topology by defining a metric on it. For elements $V, W \in \mathbb{G}E$, the *gap distance* $d_{gap}(V, W)$ between V and W is defined as the Haudorff distance of their respective closed unit balls $\overline{\mathbb{B}}_V$ and $\overline{\mathbb{B}}_W$. More precisely that means

$$d_{\mathsf{gap}}(\mathbf{V},\mathbf{W}) = d_{\mathsf{H}}(\overline{\mathbb{B}}_{\mathbf{V}},\overline{\mathbb{B}}_{\mathbf{W}}) = \max\left\{\sup_{v\in\overline{\mathbb{B}}_{\mathbf{V}}} d(v,\overline{\mathbb{B}}_{\mathbf{W}}), \sup_{w\in\overline{\mathbb{B}}_{\mathbf{W}}} d(w,\overline{\mathbb{B}}_{\mathbf{V}})\right\},$$
(A.5.3.1)

where, as usual, $d(v, B) = \inf_{w \in B} ||v - w||$ denotes the *distance* between a point $v \in E$ and a closed $B \subset E$.

5.3.2 Lemma Let

$$\vec{d}_{\mathsf{gap}}(\mathbf{V},\mathbf{W}) = \sup_{v \in \overline{\mathbb{B}}_{\mathbf{V}}} d(v, \overline{\mathbb{B}}_{\mathbf{W}})$$

denote the directed or one-sided gap between $V, W \in \mathbb{GE}$. Then the following holds true.

(i) $\vec{d}_{gap}(0, V) = \vec{d}_{gap}(V, 0) = 1$ whenever $V \neq 0$.

- (ii) $\vec{d}_{gap}(V, W) = 0$ if and only if $V \subset W$.
- (iii) For all $x \in E$,

$$d(x, \overline{\mathbb{B}}_{\mathcal{V}}) \leq d(x, \overline{\mathbb{B}}_{\mathcal{W}}) + d_{\mathsf{gap}}(\mathcal{W}, \mathcal{V}) .$$

Proof. (i) follows immediately by definition and (ii) holds true since $d(v, \overline{\mathbb{B}}_W) = 0$ if and only if $v \in \overline{\mathbb{B}}_W$. It remains to show (iii). To this end let $x \in E$, $v \in \overline{\mathbb{B}}_V$ and $w \in \overline{\mathbb{B}}_W$. Then, by the triangle inequality for the distance $d : E \times E \to \mathbb{R}$, $(x, y) \mapsto ||x - y||$,

$$d(x,v) \leq d(x,w) + d(w,v) .$$

This entails, by taking the infimum with respect to $v \in \overline{\mathbb{B}}_{V}$,

$$d(x,\overline{\mathbb{B}}_{\mathcal{V}}) \leqslant d(x,w) + d(w,\overline{\mathbb{B}}_{\mathcal{V}}) \leqslant d(x,w) + d_{\mathsf{gap}}(\mathcal{W},\mathcal{V}) \ .$$

Since $w \in \overline{\mathbb{B}}_W$ was arbitrary, (iii) follows.

5.3.3 Proposition The gap distance on the Graßmannian \mathbb{GE} of a Banach space is a metric.

Proof. By definition, the gap distance is symmetric. By (ii) of Lemma 5.3.2, one has $d_{gap}(V, W) = 0$ if and only if V = W. It remains to show the triangle inequality. Let $V, W, X \in \mathbb{GH}$ and use (iii) in the preceding lemma to verify

$$\begin{split} \vec{d}_{\mathsf{gap}}(\mathbf{X},\mathbf{V}) &= \sup_{x \in \overline{\mathbb{B}}_{\mathbf{X}}} d(x, \overline{\mathbb{B}}_{\mathbf{V}}) \leqslant \sup_{x \in \overline{\mathbb{B}}_{\mathbf{X}}} d(x, \overline{\mathbb{B}}_{\mathbf{W}}) + \vec{d}_{\mathsf{gap}}(\mathbf{W}, \mathbf{V}) \leqslant \vec{d}_{\mathsf{gap}}(\mathbf{X}, \mathbf{W}) + \vec{d}_{\mathsf{gap}}(\mathbf{W}, \mathbf{V}) ,\\ \vec{d}_{\mathsf{gap}}(\mathbf{V}, \mathbf{X}) &= \sup_{v \in \overline{\mathbb{B}}_{\mathbf{V}}} d(v, \overline{\mathbb{B}}_{\mathbf{X}}) \leqslant \sup_{v \in \overline{\mathbb{B}}_{\mathbf{V}}} d(v, \overline{\mathbb{B}}_{\mathbf{W}}) + \vec{d}_{\mathsf{gap}}(\mathbf{W}, \mathbf{X}) \leqslant \vec{d}_{\mathsf{gap}}(\mathbf{V}, \mathbf{W}) + \vec{d}_{\mathsf{gap}}(\mathbf{W}, \mathbf{X}) . \end{split}$$

This entails the triangle inequality for d_{gap} .

5.3.4 Recall that to every closed linear subspace $V \subset \mathcal{H}$ of a Hilbert space \mathcal{H} there exists a unique orthogonal projection $P_V : \mathcal{H} \to \mathcal{H}$ whose image is V. The kernel of the projection P_V coincides with the orthogonal complement V^{\perp} . One thus obtains a canonical embedding of $\mathbb{GH} \hookrightarrow \mathfrak{B}(\mathcal{H})$. The restriction of the operator norm distance to \mathbb{GH} endows \mathbb{GH} with another metric which we denote by δ .

5.3.5 Proposition ((Akhiezer & Glazman, 1993, Sec. 34)) For every Hilbert space $\mathcal H$ the metric

$$\delta : \mathbb{GH} \times \mathbb{GH} \to \mathbb{R}, \ (V, W) \mapsto \|P_V - P_W\|$$

coincides with the gap metric $d_{gap} : \mathbb{GH} \times \mathbb{GH} \to \mathbb{R}$. Moreoever, for all $V, W \in \mathbb{GH}$,

- (i) $d_{gap}(V, W) \leq 1$,
- (ii) $\vec{d}_{gap}(V, W) = ||(I P_W)P_V||$, and
- (iii) $d_{gap}(V, W) = \max \{ \| (I P_W) P_V \|, \| (I P_V) P_W \| \}$.

Proof. First note that

$$\|(I - P_{\mathbf{W}})P_{\mathbf{V}}\| = \sup_{v \in \overline{\mathbb{B}}_{\mathbf{V}}} \|v - P_{\mathbf{W}}v\| = \vec{d}_{\mathsf{gap}}(\mathbf{V}, \mathbf{W})$$

since $d(v, W) = ||v - P_W v||$ for all $v \in \overline{\mathbb{B}}_V$ by the orthogonal decomposition theorem, 3.2.3. This proves (ii) and (iii). Next observe that

$$P_{\rm V} - P_{\rm W} = P_{\rm V}(I - P_{\rm W}) - (I - P_{\rm V})P_{\rm W}$$
.

By orthogonality of the images of $P_V(I-P_W)$ and $(I-P_V)P_W$ this implies for all $x \in \mathfrak{H}$

$$\|(P_{\rm V} - P_{\rm W})x\|^{2} = \|P_{\rm V}(I - P_{\rm W})x\|^{2} + \|(I - P_{\rm V})P_{\rm W}x\|^{2} \le \\ \le \|(I - P_{\rm W})x\|^{2} + \|P_{\rm W}x\|^{2} = \|x\|^{2} ,$$
(A.5.3.2)

hence

$$\delta(\mathbf{V}, \mathbf{W}) = \|P_{\mathbf{V}} - P_{\mathbf{W}}\| \le 1$$
 (A.5.3.3)

One also obtains

$$\delta(\mathbf{V}, \mathbf{W}) = \sup_{x \in \overline{\mathbb{B}}_{\mathcal{H}}} \|(P_{\mathbf{V}} - P_{\mathbf{W}})x\| = \sup_{x \in \overline{\mathbb{B}}_{\mathcal{H}}} \sqrt{\|P_{\mathbf{V}}(I - P_{\mathbf{W}})x\|^2 + \|(I - P_{\mathbf{V}})P_{\mathbf{W}}x\|^2}.$$
 (A.5.3.4)

By restricting x to the closed ball of W this formula entails

$$\delta(\mathbf{V},\mathbf{W}) \ge \sup_{x \in \overline{\mathbb{B}}_{\mathbf{W}}} \|(I - P_{\mathbf{V}})P_{\mathbf{W}}x\| = \sup_{x \in \overline{\mathbb{B}}_{\mathbf{W}}} \|(I - P_{\mathbf{V}})x\| = \vec{d}_{\mathsf{gap}}(\mathbf{V},\mathbf{W}) \ .$$

By switching V and W in (A.5.3.3) one gets

$$\delta(\mathbf{V},\mathbf{W}) \ge \sup_{x \in \overline{\mathbb{B}}_{\mathbf{V}}} \|(I - P_{\mathbf{W}})P_{\mathbf{V}}x\| = \sup_{x \in \overline{\mathbb{B}}_{\mathbf{V}}} \|(I - P_{\mathbf{W}})x\| = \vec{d}_{\mathsf{gap}}(\mathbf{W},\mathbf{V}) \ .$$

Consequently,

$$\delta(\mathbf{V}, \mathbf{W}) \ge d_{\mathsf{gap}}(\mathbf{V}, \mathbf{W}) . \tag{A.5.3.5}$$

Let us show that

$$\delta(\mathbf{V}, \mathbf{W}) \leqslant d_{\mathsf{gap}}(\mathbf{V}, \mathbf{W}) . \tag{A.5.3.6}$$

To this end observe that for all $x\in\overline{\mathbb{B}}_{\mathcal{H}}$ by (ii) and $P^2_{\mathrm{W}}=P_{\mathrm{W}}$

$$||(I - P_{\rm V})P_{\rm W}x|| \le \vec{d}_{\sf gap}({\rm W}, {\rm V}) \cdot ||P_{\rm W}x||$$
 (A.5.3.7)

Moreover,

$$\begin{split} \|P_{\rm V}(I - P_{\rm W})x\|^2 &= \langle P_{\rm V}(I - P_{\rm W})x, P_{\rm V}(I - P_{\rm W})x \rangle = \langle P_{\rm V}^2(I - P_{\rm W})x, (I - P_{\rm W})^2x \rangle = \\ &= \langle (I - P_{\rm W})P_{\rm V}^2(I - P_{\rm W})x, (I - P_{\rm W})x \rangle \leqslant \\ &\leqslant \vec{d}_{\sf gap}({\rm V}, {\rm W}) \|P_{\rm V}(I - P_{\rm W})x\| \, \|(I - P_{\rm W})x\| \, , \end{split}$$

 \rightarrow

and therefore

$$||P_{\rm V}(I - P_{\rm W})x|| \le d_{\sf gap}({\rm V}, {\rm W}) ||(I - P_{\rm W})x||$$
 . (A.5.3.8)

Inserting this estimate and (A.5.3.7) into the squared right side of (A.5.3.4) then gives

$$\begin{aligned} \|P_{\mathcal{V}}(I-P_{\mathcal{W}})x\|^{2} + \|(I-P_{\mathcal{V}})P_{\mathcal{W}}x\|^{2} &\leq \vec{d}_{\mathsf{gap}}^{2}(\mathcal{W},\mathcal{V}) \cdot \|P_{\mathcal{W}}x\|^{2} + \vec{d}_{\mathsf{gap}}^{2}(\mathcal{V},\mathcal{W}) \,\|(I-P_{\mathcal{W}})x\|^{2} &\leq \\ &\leq d_{\mathsf{gap}}^{2}(\mathcal{W},\mathcal{V}) \cdot \left(\|P_{\mathcal{W}}x\|^{2} + \|(I-P_{\mathcal{W}})x\|^{2}\right) = d_{\mathsf{gap}}^{2} \,\|x\|^{2} \,. \end{aligned}$$

Comparing with the left side of (A.5.3.4) shows (A.5.3.6), and the equality of δ and d_{gap} follows. By (A.5.3.3) the latter also yields (i).

5.3.6 Remark We will use the symbols d_{gap} and δ interchangeably to denote the gap metric on the Graßmannian of a Banach space .

5.3.7 Theorem Eqipped with the gap metric the Graßmann manifold of a Banach space is a complete metric space.

Proof. We present the proof for the underlying Banach space being a Hilbert space \mathcal{H} . Then the claim followsimmediately from the fact that $\mathfrak{B}(\mathcal{H})$ is complete and that the limit of a Cauchy sequence of orthogonal projections $(P_n)_{n\in\mathbb{N}} \subset \mathfrak{B}(\mathcal{H})$ is again an orthogonal projection. The general case is more tricky.

A.6. Lie groups

A.6.1. Symmetry groups of bilinear and sesquilinear forms

6.1.1 In this section \mathbb{K} will always stand for the field of real or complex numbers. Before defining their symmetry groups let us recall the notions of bilinear and sesquilinear forms. A *bilinear form* on a \mathbb{K} -vector space V is a map $b : \mathbb{V} \times \mathbb{V} \to \mathbb{K}$ having the following properties:

(BF1) The map b is *linear* in its first coordinate which means that

$$b(v_1 + v_2, w) = b(v_1, w) + b(v_2, w)$$
 and $b(rv, w) = rb(v, w)$

for all $v, v_1, v_2, w \in V$ and $r \in \mathbb{K}$.

(BF2), (SF2) The map b is *linear* in its second coordinate which means that

$$b(v, w_1 + w_2) = b(v, w_1) + b(v, w_2)$$
 and $b(v, rw) = rb(v, w)$

for all $v, w, w_1, w_2 \in V$ and $r \in \mathbb{K}$.

Bilinear forms with the property that commuting its variables leads to the same or to the negative of the original bilinear form are given a particular name. More precisely, a bilinear map $b: V \times V \rightarrow \mathbb{K}$ is said to be *symmetric* if

(BF3s) b(v, w) = b(w, v) for all $v, w \in V$,

and antisymmetric or skew-symmetric if

(BF3a) b(v, w) = -b(w, v) for all $v, w \in V$.

A map $b : V \times V \rightarrow \mathbb{K}$ which satisfies (BF2) and axiom (SF1) below instead of (BF1) is called a *sesquilinear form*.

(SF1) The map b is conjugate-linear in its first coordinate which means that

$$b(v_1 + v_2, w) = b(v_1, w) + b(v_2, w)$$
 and $b(rv, w) = \overline{r}b(v, w)$

for all $v, v_1, v_2, w \in V$ and $r \in \mathbb{K}$.

A sesquilinear form b is called a *hermitian form* if it has the following property:

(SF3c) The map b is conjugate-symmetric which means that

$$b(v,w) = b(w,v)$$
 for all $v,w \in V$.

If the ground field of the underlying vector space of a bilinear or sesquilinear form b is \mathbb{C} , one calls b a *complex* bilinear form respectively a *complex* sesquilinear form. One uses analogous language when the ground field is \mathbb{R} . Note that a real sesquilinear form is the same as a real bilinear form.

A bilinear or sesquilinear form b is said to be *weakly-nondegenerate* if it satisfies axiom

(SF4w) The map $^{\flat}: V \to V'$, $v \mapsto v^{\flat} = b(-, v) = (V \ni w \to b(w, v) \in \mathbb{K})$ from V to its algebraic dual V' is injective .

Note that (SF4w) is equivalent to the requirement that for every $v \in V$ the map $b(v, -) : V \to \mathbb{K}$, $w \to b(v, w)$ is the zero map if and only if v = 0.

In case the underlying vector space V is normed, there is a stronger version of nondegeneracy for bounded bilinear or sesquilinear forms $b : V \times V \rightarrow \mathbb{K}$. Namely, one calls such a form *nondegenerate* if it fulfills

(SF4n) The map $^{\flat}: V \to V^*$, $v \mapsto v^{\flat} = b(v, -) = (V \ni w \to b(v, w) \in \mathbb{K})$ from V to its topological dual V^{*} is a linear or conjugate-linear topological isomorphism.

Recall that $b(v, v) \in \mathbb{R}$ for every hermitian form b on V and $v \in V$. In case that such a b satisfies

(SF5s)
$$b(v,v) \ge 0$$
 for all $v \in V$,

then one calls the hermitian form *b* positive semidefinite.

Recall from Lemma 3.1.6 that a positive semidefinite hermitian form b on a \mathbb{K} -vector space V is weakly-nondegenerate if and only if it is *positive definite* which means that

(SF5p) b(v, v) > 0 for all $v \in V \setminus \{0\}$.

6.1.2 Remark If b is a nondegenerate bilinear or sesquilinear form on a Banach space V, then one sometimes calls the map ${}^{\flat}$: V \rightarrow V* from Axiom (SF4n) and its inverse ${}^{\sharp}$: V* \rightarrow V the *musical isomorphisms* associated to b.

6.1.3 Examples In addition to the hermitian forms introduced in Examples 3.1.9 let us give a few more examples of bilinear forms which are particularly relevant for mathematics or mathematical physics.

(a) Let p, q be positive integers, n = p + q, and $\langle \cdot, \cdot \rangle_{p,q} : \mathbb{R}^n \times \mathbb{R}^n \to \mathbb{R}$ the *pseudo-euclidean* form given by

$$\langle x, y \rangle_{p,q} = \sum_{i=1}^{p} x^{i} y^{i} - \sum_{j=p+1}^{n} x^{j} y^{j}$$
 for $x = (x^{1}, \dots, x^{n}), y = (y^{1}, \dots, y^{n}) \in \mathbb{R}^{n}$

The map $\langle \cdot, \cdot \rangle_{p,q}$ is a nondegenerate bilinear form, but it is not positive semidefinite by definition. The space \mathbb{R}^n together with the pseudo-euclidean form $\langle \cdot, \cdot \rangle_{p,q}$ is sometimes denoted $\mathbb{R}^{p,q}$. For the particular case (p,q) = (1,d) = (1,n-1) one calls $\mathbb{R}^{1,d}$ *Minkowski space* of *space-time dimension* d + 1, and $\langle \cdot, \cdot \rangle_{\mathsf{M}} := \langle \cdot, \cdot \rangle_{1,d}$ the corresponding *Minkowski metric*. The components of elements $x, y \in \mathbb{R}^{1,d}$ of Minkowski space are often indexed in the form $x = (x^0, x^1, \ldots, x^d) = (x^{\mu})_{\mu=0}^d$ and $y = (y^0, y^1, \ldots, y^d) = (y^{\nu})_{\nu=0}^d$. In this notation, the Minkowski metric is given by

$$\langle x, y \rangle_{\mathsf{M}} = x^0 y^0 - \sum_{i=1}^d x^i y^i \; .$$

Moreover, one associates to x and y the *space-vectors*

$$\vec{x} = (x^1, \dots, x^d) = (x^i)_{i=1}^d$$
 and $\vec{y} = (y^1, \dots, y^d) = (y^j)_{j=1}^d$.

When labels run through all space-time indices they are usually denoted in the mathematical physics literature by lower-case Greek letters, when they run only through space indices, they are denoted by lower-case Roman letters. We will follow these conventions.

(b) Next consider \mathbb{K}^{2n} with $n \in \mathbb{N}_{>0}$ and define

$$\omega: \mathbb{K}^{2n} \times \mathbb{K}^{2n} \to \mathbb{K}, \quad (v, w) \mapsto \sum_{i=1}^{n} (v_i \, w_{n+i} - w_i \, v_{n+i})$$

Then ω is a nondegenerate antisymmetric bilinear form. We call it the *standard symplectic form* on \mathbb{K}^{2n} . More generally, a nondegenerate antisymmetric bilinear form $\omega : E \times E \to \mathbb{K}$ on a Banach space E over \mathbb{K} is called a *symplectic form*. If $\omega : E \times E \to \mathbb{K}$ is only weakly-nondegenerate (but still antisymmetric), then one says that ω is a weakly-symplectic form.

If V is a Banach space and $E=V\oplus V^{\ast}\text{, then}$

$$\omega : \mathbf{E} \times \mathbf{E} \to \mathbb{K}, \ ((v, \alpha), (w, \beta)) \mapsto \beta(v) - \alpha(w)$$

is a weakly-symplectic form on E which is symplectic if and only if V is reflexive that is if and only if the canonical embedding $V \hookrightarrow V^{**}$ is an isomorphism.

Proof. Antisymmetry is clear by definition.

6.1.4 Next consider a Banach space E over \mathbb{K} with norm $\|\cdot\|$ and the space $\mathfrak{B}(E)$ of bounded \mathbb{K} -linear operators on E. Recall that $\mathfrak{B}(E)$ carries the following natural topologies:

(i) the norm topology or uniform operator topology \mathcal{T}_n defined by the operator norm

$$\|-\|:\mathfrak{B}(\mathcal{E})\to\mathbb{R}_{\geq 0},\ A\mapsto\|A\|:=\sup\left\{\|Av\|\in\mathbb{R}_{\geq 0}\mid v\in\mathcal{E}\ \&\ \|v\|\leqslant 1\right\}\,,$$

(ii) the compact-open topology T_{co} defined by the seminorms

$$p_K : \mathfrak{B}(\mathbf{E}) \to \mathbb{R}_{\geq 0}, A \mapsto p_K(A) := \sup \{ \|Av\| \in \mathbb{R}_{\geq 0} \mid v \in K \},\$$

where K runs through the nonempty compact subsets of E,

(iii) the strong operator topology T_s defined by the seminorms

$$p_v: \mathfrak{B}(\mathbf{E}) \to \mathbb{R}_{\geq 0}, A \mapsto p_v(A) := ||Av||,$$

where v runs through the nonzero elements of E,

(iv) the weak operator topology T_w defined by the seminorms

$$p_{\lambda,v}: \mathfrak{B}(\mathbf{E}) \to \mathbb{R}_{\geq 0}, A \mapsto p_{\lambda,v}(A) := \lambda(Av) ,$$

where λ runs through the nonzero bounded linear functionals $E \to \mathbb{K}$ and v through the nonzero elements of E.

These four operator topologies are comparable. More precisely one has

$$\mathfrak{T}_{\mathsf{w}} \subset \mathfrak{T}_{\mathsf{s}} \subset \mathfrak{T}_{\mathsf{co}} \subset \mathfrak{T}_{\mathsf{n}}$$
 .

In case E is finite dimensional, the topologies coincide, if E is infinite dimensional, then the inclusions are proper.

To denote which topology $\mathfrak{B}(E)$ is endowed with we write $\mathfrak{B}(E)_n$, $\mathfrak{B}(E)_{co}$, $\mathfrak{B}(E)_s$ and $\mathfrak{B}(E)_w$, respectively.

6.1.5 Proposition and Definition Let E be a Banach space over \mathbb{K} , and $GL(E) \subset \mathfrak{B}(E)_n$ the space of bounded invertible \mathbb{K} -linear operators on E endowed with the norm topology. Then the following holds true.

- (a) The space GL(E) is open in $\mathfrak{B}(E)_n$.
- (b) GL(E) together with the operator product and the identity map id_E is a group.
- (c) The group G := GL(E) endowed with the norm topology is a topological group which means that it has the following properties:

(TopGr1) The multiplication map $\cdot : G \times G \rightarrow G$ is continuous.

(TopGr2) The inversion map $i : G \rightarrow G$ is continuous.

Proof. ad (b). By the open mapping theorem the inverse of a bounded invertible operator is bounded as well, hence $g^{-1} \in GL(E)$ for all $g \in GL(E)$. Obviously $id_E \in GL(E)$, so GL(E) is a group indeed.

ad (a). Let $g \in GL(E)$. Then $||g^{-1}|| > 0$, since $1 = ||v|| \le ||g^{-1}|| ||gv||$ for every unit vector $v \in E$. Let $0 < r < \frac{1}{||g^{-1}||}$. For $A \in \mathfrak{B}(E)$ with ||A|| < r the series $\sum_{k \in \mathbb{N}} (-1)^k (g^{-1}A)^k$ then is dominated by the converging geometric series $\sum_{k \in \mathbb{N}} r^k$, hence converges too. Compute

$$\left(\mathrm{id}_{\mathrm{E}} + g^{-1}A\right)\left(\sum_{k=0}^{\infty} (-1)^{k} \left(g^{-1}A\right)^{k}\right) = \sum_{k=0}^{\infty} (-1)^{k} \left(g^{-1}A\right)^{k} - \sum_{k=1}^{\infty} (-1)^{k} \left(g^{-1}A\right)^{k} = \mathrm{id}_{\mathrm{E}}$$

and analogously

$$\left(\sum_{k=0}^{\infty} (-1)^k (g^{-1}A)^k\right) (\mathrm{id}_{\mathrm{E}} + g^{-1}A) = \mathrm{id}_{\mathrm{E}} .$$

Therefore $id_E + g^{-1}A$ is invertible with bounded inverse $\sum_{k=0}^{\infty} (-1)^k (g^{-1}A)^k$. Hence the operator $g + A = g(id_E + g^{-1}A)$ is invertible as well and the open ball of radius r around g is contained in GL(E). Thus GL(E) is open in $\mathfrak{B}(E)$.

ad (c). To verify continuity of multiplication recall that $||AB|| \leq ||A|| ||B||$ for all $A, B \in \mathfrak{B}(E)$. Then

$$||AB - A'B'|| = ||(AB - A'B) + (A'B - A'B')|| \le ||A - A'|| ||B|| + ||A'|| ||B - B'||.$$

Hence multiplication is locally Lipschitz continuous, therefore continuous.

To prove continuity of inversion let $g \in GL(E)$ and choose $0 < r < \frac{1}{\|g^{-1}\|}$. Then $g + A \in GL(E)$ for all $A \in \mathfrak{B}(E)$ with $\|A\| < r$ by the preceding considerations. Moreover,

$$\begin{split} \left\| (g+A)^{-1} - g^{-1} \right\| &= \left\| \left((\mathrm{id}_{\mathrm{E}} + g^{-1}A)^{-1} - \mathrm{id}_{\mathrm{E}} \right) g^{-1} \right\| \leqslant \\ &\leq \left\| g^{-1} \right\| \, \left\| \sum_{k=1}^{\infty} (-1)^k \big(g^{-1}A \big)^k \right\| \leqslant \left\| g^{-1} \right\| \, \sum_{k=1}^{\infty} \left\| g^{-1}A \right\|^k \leqslant \frac{\left\| g^{-1} \right\|^2}{1 - r \|g^{-1}\|} \|A\| \; . \end{split}$$

Hence inversion is locally Lipschitz continuous, so in particular continuous.

Unless mentioned differently, we assume from now on that GL(E) carries the norm topology. Sometimes we will write $GL(E)_n$ to emphasize this.

6.1.6 Assume that $b : E \times E \to \mathbb{K}$ is a bounded bilinear or sesquilinear form on a Banach space E over \mathbb{K} . Consider the group GL(E) and define G(E, b) as the set of all $g \in GL(E)$ such that

$$b(gv, gw) = b(v, w)$$
 for all $v, w \in E$.

6.1.7 Proposition Under the assumptions stated G(E, b) is a closed subgroup of $GL(E)_n$.

Proof. If $g, h \in G(E, b)$, then their operator product gh lies in G(E, b) as well since

$$b(ghv, ghw) = b(hv, hw) = b(v, w)$$
 for all $v, w \in E$.

Moreover, id_E leaves b invariant, so is in G(E, b), too. Hence G(E, b) is a subgroup of $GL(E)_n$.

Now assume that $g \in GL(E)_n \setminus G(E, b)$. Then there are $v, w \in E$ such that

$$b(gv, gw) \neq b(v, w)$$
 and $||v|| = ||w|| = 1$.

Put $\delta = |b(gv, gw) - b(v, w)|$ and let $C = \sup \{|b(x, y)| \mid x, y \in E \& ||x|| = ||y|| = 1\}$. Then one has for all $h \in \mathfrak{B}(E)$

$$\begin{aligned} |b(hv, hw) - b(v, w)| &= \left| \left(b(hv, hw) - b(gv, gw) \right) - \left(b(v, w) - b(gv, gw) \right) \right| \geqslant \\ &\geqslant \left| \delta - |b(hv, hw) - b(gv, gw)| \right| \geqslant \\ &\geqslant \delta - |b(hv, hw) - b(gv, hw)| - |b(gv, hw) - b(gv, gw)| \geqslant \\ &\geqslant \delta - C \left\| h - g \right\| \left(\left\| h \right\| + \left\| g \right\| \right) . \end{aligned}$$

Hence, if $||h - g|| < \varepsilon$ with $\varepsilon = \min\left\{1, \frac{1}{2||g^{-1}||}, \frac{\delta}{2(C+1)(2||g||+1)}\right\}$, then $h \in \mathsf{GL}(E)$ and $|b(hv, hw) - b(v, w)| \ge \delta - C\left(2||g|| + 1\right)\varepsilon \ge \frac{1}{2}\delta$.

So $GL(E)_n \setminus G(E, b)$ is open and the claim is proved.

6.1.8 Examples (a) For a Hilbert space H, the group G(H, ⟨·, ·⟩) is called the *unitary group* of H and denoted U(H). If the underlying ground field is R, one often writes O(H) for G(H, ⟨·, ·⟩) and calls it the *orthogonal group* of the real Hilbert space H. In the finite dimensional case, U(n) stands for U(Cⁿ) and O(n) for O(Rⁿ).

- (b) Given two positive integers p, q consider the pseudo-euclidean metric $\langle \cdot, \cdot \rangle_{p,q}$ on $\mathbb{R}^{p,q} \cong \mathbb{R}^{p+q}$, see Example 6.1.3 (a). The invariant group $G(\mathbb{R}^n, \langle \cdot, \cdot \rangle_{p,q})$ then is called a *pseudo-orthogonal* group and is denoted O(p,q).
- (c) Let E be a Banach space over \mathbb{K} with a symplectic form ω . The group $Sp(E, \omega) := G(E, \omega)$ then is the symplectic group associated to (E, ω) . If E is \mathbb{K}^{2n} and ω its canonical symplectic form, then one writes $Sp(2n, \mathbb{K})$ for the associated symplectic group.

6.1.9 It has been claimed wrongly at several places in the mathematical literature, notably in Simms (1968)[Proof of Thm. 1] and Atiyah & Segal (2004)[p. 321] that the unitary group with the strong operator topology respectively with the compact-open topology is not a topological group. The correct(ed) statement appeared in Schottenloher (1995)[III.3.2 Satz], Neeb (1997)[Prop. II.1], and Schottenloher (2008)[Prop.3.11], whose presentation we will essentially follow here.

6.1.10 Proposition If \mathcal{H} is a Hilbert space, then $U(\mathcal{H})_s$, the unitary group $U(\mathcal{H})$ endowed with the strong operator topology, is a complete topological group. Moreover, the compact-open topology, the strong operator topology, and weak operator topology all coincide on $U(\mathcal{H})$. Finally, if \mathcal{H} is separable, then $U(\mathcal{H})_s$ is completely metrizable.

Proof. For $v \in \mathcal{H}$ and $V \in U(\mathcal{H})$ let $p_{v,V} : U(\mathcal{H}) \to \mathbb{R}_{\geq 0}$ be defined by

$$U \mapsto p_{v,V}(U) = \|(U - V)v\|$$

A subbasis of the strong operator topology on $U(\mathcal{H})$ then is given by the sets

$$\left\{ U \in \mathsf{U}(\mathcal{H}) \mid p_{v,V}(U) < \varepsilon \right\}, \quad \text{where } v \in \mathcal{H}, \ V \in \mathsf{U}(\mathcal{H}), \text{ and } \varepsilon > 0 \ .$$

A.6.2. The Lie group SO(3) and its universal cover SU(2)

6.2.1 Recall that the *orthogonal group* in real dimension 3 is given by

$$\mathsf{O}(3) = \left\{ g \in \mathsf{GL}(3,\mathbb{R}) \mid \forall \vec{x}, \vec{y} \in \mathbb{R}^3 : \langle g\vec{x}, g\vec{y} \rangle = \langle \vec{x}, \vec{y} \rangle \right\} \,.$$

The special orthogonal group in dimension 3 is the subgroup

$$SO(3) = \{g \in O(3) \mid \det g = 1\}$$
.

Let us show that both are Lie groups. Consider the map

$$f: \mathsf{GL}(3,\mathbb{R}) \to \mathfrak{Sym}(3,\mathbb{R}), \quad g \mapsto g^{\mathrm{t}}g$$

where g^t is the transpose of g and $\mathfrak{Sym}(n,\mathbb{R})$ denotes the space of real symmetric $n \times n$ matrices. Note that $\dim \mathfrak{Sym}(n,\mathbb{R}) = \frac{n(n+1)}{2}$ and that f is well-defined since $(g^tg)^t = g^tg$. We show that f is a submersion. To this end check first that for every $g \in \mathsf{GL}(3,\mathbb{R})$ the tangent map of f at g is

$$T_g f : \mathfrak{gl}(3,\mathbb{R}) \to \mathfrak{Sym}(3,\mathbb{R}), \quad A \mapsto A^{\mathrm{t}}g + g^{\mathrm{t}}A.$$

For given $S \in \mathfrak{Sym}(3,\mathbb{R})$ put $A = \frac{1}{2} (g^t)^{-1} S$ and compute

$$T_g f(A) = \frac{1}{2} \left(S^{\mathsf{t}} + S \right) = S \; .$$

Hence f is a submersion, and $O(3) = f^{-1}(I_3)$ is a submanifold of $GL(3, \mathbb{R})$ of dimension $\dim_{\mathbb{R}} GL(3, \mathbb{R}) - \dim_{\mathbb{R}} \mathfrak{Sym}(3, \mathbb{R}) = 9 - 6 = 3$. Because the group multiplication and inverse on $GL(3, \mathbb{R})$ are smooth, their restriction to O(3) is so, too, and O(3) is a Lie group. Since $(\det g)^2 = \det g \det g^t = 1$ for all $g \in O(3)$, the subgroup $SO(3) = O(3) \cap \det^{-1}(\mathbb{R}_{>0})$ is open in O(3), and O(3) is the disjoint union of SO(3) and -SO(3). Moreover, SO(3) becomes a Lie group.

The Lie algebra $\mathfrak{o}(3)$ of O(3) coincides with the Lie algebra $\mathfrak{so}(3)$ of SO(3) and can be determined via the submersion f, too. More precisely

$$\mathfrak{o}(3) = \mathfrak{so}(3) = \ker T_{\mathbb{I}} f = \left\{ A \in \mathfrak{gl}(3, \mathbb{R}) \mid A^{\mathrm{t}} + A = 0 \right\} \,,$$

and $\mathfrak{so}(3)$ is the space of all skew-symmetric real 3×3 matrices. Note that $\operatorname{tr} A = 0$ for every element $A \in \mathfrak{so}(3)$.

6.2.2 Theorem The Lie algebras (\mathbb{R}^3, \times) and $\mathfrak{so}(3)$ are isomorphic. An isomorphism is given by the map

$$M: \mathbb{R}^3 \to \mathfrak{so}(3), \quad \vec{x} = \begin{pmatrix} x^1 \\ x^2 \\ x^3 \end{pmatrix} \mapsto M_{\vec{x}} = \begin{pmatrix} 0 & -x^3 & x^2 \\ x^3 & 0 & -x^1 \\ -x^2 & x^1 & 0 \end{pmatrix}$$

Denoting by $\vec{e}_1, \vec{e}_2, \vec{e}_3$ the standard basis of \mathbb{R}^3 , the elements

$$J_x = J_1 = M_{\vec{e}_1} = \begin{pmatrix} 0 & 0 & 0 \\ 0 & 0 & -1 \\ 0 & 1 & 0 \end{pmatrix},$$
$$J_y = J_2 = M_{\vec{e}_2} = \begin{pmatrix} 0 & 0 & 1 \\ 0 & 0 & 0 \\ -1 & 0 & 0 \end{pmatrix},$$
$$J_z = J_3 = M_{\vec{e}_3} = \begin{pmatrix} 0 & -1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 0 \end{pmatrix}$$

form a basis of the Lie algebra $\mathfrak{so}(3)$. These elements are sometimes called the (standard) infinitesimal generators of rotations.

Proof. By definition M is linear. Moreover, the images $M_{\vec{e}_k}$, k = 1, 2, 3, are linearly independent, so by dimension reasons the map M is a linear isomorphism. It remains to show that M preserves the Lie brackets. To this end compute for $\vec{x}, \vec{y} \in \mathbb{R}^3$

$$\vec{x} \times \vec{y} = \begin{pmatrix} x^1 \\ x^2 \\ x^3 \end{pmatrix} \times \begin{pmatrix} y^1 \\ y^2 \\ y^3 \end{pmatrix} = \begin{pmatrix} x^2 y^3 - x^3 y^2 \\ x^3 y^1 - x^1 y^3 \\ x^1 y^2 - x^2 y^1 \end{pmatrix},$$

and then

$$M_{\vec{x}} \cdot M_{\vec{y}} = \begin{pmatrix} 0 & -x^3 & x^2 \\ x^3 & 0 & -x^1 \\ -x^2 & x^1 & 0 \end{pmatrix} \cdot \begin{pmatrix} 0 & -y^3 & y^2 \\ y^3 & 0 & -y^1 \\ -y^2 & y^1 & 0 \end{pmatrix} = \\ = \begin{pmatrix} -x^3 y^3 - x^2 y^2 & x^2 y^1 & x^3 y^1 \\ x^1 y^2 & -x^3 y^3 - x^1 y^1 & x^3 y^2 \\ x^1 y^3 & x^2 y^3 & -x^2 y^2 - x^1 y^1 \end{pmatrix}$$

Forming the commutator gives

$$\begin{bmatrix} M_{\vec{x}}, M_{\vec{y}} \end{bmatrix} = \begin{pmatrix} 0 & x^2 y^1 - x^1 y^2 & x^3 y^1 - x^1 y^3 \\ x^1 y^2 - x^2 y^1 & 0 & x^3 y^2 - x^2 y^3 \\ x^1 y^3 - x^3 y^1 & x^2 y^3 - x^3 y^2 & 0 \end{pmatrix} = M_{\vec{x} \times \vec{y}} .$$

Hence ${\cal M}$ preserves Lie brackets and the claim is proved.

6.2.3 Now let us consider the special unitary group

$$\mathsf{SU}(2) = \left\{ g \in \mathsf{GL}(2,\mathbb{C}) \mid \forall v, w \in \mathbb{C}^2 : \langle gv, gw \rangle = \langle v, w \rangle \& \det g = 1 \right\}.$$

To verify that SU(2) is a Lie group let f be the map

$$f: \mathsf{GL}(2,\mathbb{C}) \to \mathfrak{Herm}(2), \quad g \mapsto g^*g \;,$$

where $\mathfrak{Herm}(n)$ denotes the space of hermitian $n \times n$ matrices. The tangent map of f at $g \in \mathsf{GL}(2,\mathbb{C})$ is given by

$$T_g f : \mathfrak{gl}(2,\mathbb{C}) \to \mathfrak{Herm}(2), \quad A \mapsto A^*g + g^*A$$
.

For given $H \in \mathfrak{Herm}(2)$ let $A = \frac{1}{2}(g^*)^{-1}H$. Then

$$T_g f(A) = \frac{1}{2} (H^* + H) = H ,$$

which entails that f is a submersion. Hence $U(2) = f^{-1}(I_2)$ is a real Lie group of dimension $\dim_{\mathbb{R}} \operatorname{GL}(2,\mathbb{C}) - \dim_{\mathbb{R}} \mathfrak{Herm}(2) = 8 - 4 = 4$. Recall that U(2) is the *unitary group* in dimension 2. The Lie algebra of U(2) is given by $\mathfrak{u}(2) = \ker T_{\mathbb{I}}f$, the space of all skew-hermitian 2×2 matrices. The determinant function det : $U(2) \to \mathbb{S}^1$ is a smooth group homomorphism and a submersion. The latter is true because for all $A \in \mathfrak{u}(2)$

$$T_{\mathbb{I}} \det(A) = \left. \frac{\partial}{\partial t} \right|_{t=0} \det(\exp(tA)) = \left. \frac{\partial}{\partial t} \right|_{t=0} e^{t \operatorname{tr} A} = \operatorname{tr} A$$

and because the matrix $A = i \mathbb{1} \in \mathfrak{gl}(2, \mathbb{C})$ is skew-hermitian and its trace $\operatorname{tr} A = 2i$ spans $\mathfrak{lie}(\mathbb{S}^1) = \mathbb{R}i$. Therefore, $\mathsf{SU}(2)$ is a real Lie group of dimension $\dim_{\mathbb{R}} \mathsf{U}(2) - \dim_{\mathbb{R}} \mathbb{R}i = 3$ and with Lie algebra $\mathfrak{su}(2)$ given by the skew-hermitian matrices of trace 0.

6.2.4 Proposition The Lie group SU(2) is homeomorphic to \mathbb{S}^3 , so in particular compact and simply connected. A homeomorphism is given by

$$\Psi: \mathbb{S}^3 \mapsto \mathsf{SU}(2), \ (x^0, x^1, x^2, x^3) \mapsto \begin{pmatrix} x^0 + x^1 \mathbf{i} & x^2 + x^3 \mathbf{i} \\ -x^2 + x^3 \mathbf{i} & x^0 - x^1 \mathbf{i} \end{pmatrix} \ .$$

Proof. One has for $(x^0,x^1,x^2,x^3)\in\mathbb{S}^3$

$$\begin{pmatrix} x^0 + x^1 \mathbf{i} & x^2 + x^3 \mathbf{i} \\ -x^2 + x^3 \mathbf{i} & x^0 - x^1 \mathbf{i} \end{pmatrix} \cdot \begin{pmatrix} x^0 - x^1 \mathbf{i} & -x^2 - x^3 \mathbf{i} \\ x^2 - x^3 \mathbf{i} & x^0 + x^1 \mathbf{i} \end{pmatrix} = \mathbb{1} \ ,$$

hence the matrix $\Psi(x^0, x^1, x^2, x^3)$ is unitary. So Ψ is well-defined. The map Ψ is obviously continuous and injective. It remains to show that Ψ is surjective, because then, by compactness of the 3-sphere, the map Ψ is a homeomorphism and SU(2) has to be compact. Let

$$g = \begin{pmatrix} z & u \\ v & w \end{pmatrix}$$

be a unitary matrix with determinant being 1 that is zw - uv = 1. By unitarity and the formula for the inverse of a 2×2 matrix one obtains the equality

$$\begin{pmatrix} w & -u \\ -v & z \end{pmatrix} = \begin{pmatrix} \overline{z} & \overline{v} \\ \overline{u} & \overline{w} \end{pmatrix} ,$$

hence $w = \overline{z}$ and $v = -\overline{u}$. Inserting this in the equation for the determinant entails that $|z|^2 + |u|^2 = 1$. Now write $z = x^0 + x^1 i$ and $u = x^2 + x^3 i$ with real x^0, x^1, x^2, x^3 . Then $(x^0, x^1, x^2, x^3) \in \mathbb{S}^3$ and $g = \Psi(x^0, x^1, x^2, x^3)$, so Ψ is surjective and the proposition is proved.

6.2.5 Proposition Consider the space

$$\mathfrak{Herm}^{\mathsf{tr}0}(2) = \mathsf{i}\,\mathfrak{su}(2) = \left\{ X \in \mathfrak{gl}(2,\mathbb{C}) \mid X^* = X \And \mathrm{tr}\, X = 0 \right\}$$

of all traceless hermitian 2×2 matrices. Then $\mathfrak{Herm}^{\mathsf{tr}0}(2)$ is a real vector space of dimension 3 with a basis given by the Pauli matrices

$$\sigma_1 = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}, \quad \sigma_2 = \begin{pmatrix} 0 & -i \\ i & 0 \end{pmatrix}, \quad \sigma_3 = \begin{pmatrix} 1 & 0 \\ 0 & -1 \end{pmatrix}$$

and with inner product

$$\langle \cdot, \cdot \rangle_{\mathfrak{h}^{\mathrm{tr0}}} : \mathfrak{Herm}^{\mathrm{tr0}}(2) \times \mathfrak{Herm}^{\mathrm{tr0}}(2) \to \mathbb{R}, \quad (X, Y) \mapsto -\frac{1}{2} \left(\det(X + Y) - \det X - \det Y \right)$$

and corresponding norm

$$\|\cdot\|_{\mathfrak{h}^{\mathrm{tr}0}} : \mathfrak{Herm}^{\mathrm{tr}0}(2) \to \mathbb{R}_{\geq 0}, \quad X \mapsto \sqrt{-\det(X)} .$$

An isometric isomorphism between $(\mathbb{R}^3, \langle \cdot, \cdot \rangle)$ and $(\mathfrak{Herm}^{tr0}(2), \langle \cdot, \cdot \rangle_{\mathfrak{h}^{tr0}})$ is given by

$$\vec{\sigma} : \mathbb{R}^3 \to \mathfrak{Herm}^{\mathsf{tr}0}(2), \quad \vec{x} \mapsto \vec{\sigma} \cdot \vec{x} = \sum_{k=1}^3 x^k \, \sigma_k \; .$$

Its inverse maps $X \in \mathfrak{Herm}^{\mathsf{tr}0}(2)$ to the vector \vec{x} with components $x^k = \frac{1}{2} \operatorname{tr}(X\sigma_k)$, where k = 1, 2, 3.

Proof. Let $X = \begin{pmatrix} a & z \\ w & d \end{pmatrix} \in \mathfrak{Herm}^{\mathsf{tr0}}(2)$. Then $a, d \in \mathbb{R}$ and $w = \overline{z}$, since X is hermitian. The assumption $\operatorname{tr} X = 0$ implies d = -a. Hence X is of the form

$$\begin{pmatrix} a & b+c i \\ b-c i & -a \end{pmatrix} = a\sigma_3 + c\sigma_1 + b\sigma_2$$

with $a, c, d \in \mathbb{R}$, and any such matrix is an element of $\mathfrak{Herm}^{\mathsf{tr0}}(2)$. Since the Pauli matrices are obviously linearly independent, they therefore form a basis of $\mathfrak{Herm}^{\mathsf{tr0}}(2)$.

Next compute for $\vec{x} = (x^1, x^2, x^3) \in \mathbb{R}^3$

$$\det\left(\vec{\sigma}\cdot\vec{x}\right) = \det\left(\begin{array}{cc} x^3 & x^2 + x^1 \,\mathrm{i} \\ x^2 - x^1 \,\mathrm{i} & -x^3 \end{array}\right) = -(x^3)^2 - (x^1)^2 - (x^2)^2 = -\|\vec{x}\|^2 \,. \tag{A.6.2.1}$$

Hence the map $\mathfrak{Herm}^{\mathrm{tr}0}(2) \to \mathbb{R}_{\geq 0}$, $X \mapsto \sqrt{-\det(X)}$ is a norm on $\mathfrak{Herm}^{\mathrm{tr}0}(2)$ which has to fullfill the parallelogram identity since the euclidean norm $\|\cdot\|$ does.

The norm on $\mathfrak{Herm}^{tr0}(2)$ is therefore induced by an inner product which can be recovered by the polarization identity (A.3.1.9) that is by

$$\langle X, Y \rangle_{\mathfrak{h}^{\mathrm{tr}0}} = -\frac{1}{2} \left(\det(X+Y) - \det X - \det Y \right) \quad \text{for all } X, Y \in \mathfrak{Herm}^{\mathrm{tr}0}(2) \;.$$

Moreover, $\vec{\sigma}$ preserves norms by (A.6.2.1), hence is an isometry.

For the remaining part of the claim check first that $(\sigma_k)^2 = 1$ for k = 1, 2, 3 and that

$$\sigma_1 \sigma_2 = \begin{pmatrix} i & 0 \\ 0 & -i \end{pmatrix} = i\sigma_3, \ \sigma_2 \sigma_3 = \begin{pmatrix} 0 & i \\ i & 0 \end{pmatrix} = i\sigma_1, \ \sigma_3 \sigma_1 = \begin{pmatrix} 0 & 1 \\ -1 & 0 \end{pmatrix} = i\sigma_2.$$
 (A.6.2.2)

Since the Pauli matrices are hermitian, forming the hermitian conjugate on both sides of these equations entails

$$\sigma_2 \sigma_1 = -i\sigma_3, \quad \sigma_3 \sigma_2 = -i\sigma_1, \quad \sigma_1 \sigma_3 = -i\sigma_2 . \tag{A.6.2.3}$$

Now compute for $\vec{x} \in \mathbb{R}^3$ and k=1,2,3

$$\frac{1}{2}\operatorname{tr}\left(\left(\vec{x}\cdot\vec{\sigma}\right)\sigma_{k}\right) = \frac{1}{2}\operatorname{tr}\left(x_{k}\left(\sigma_{k}\right)^{2}\right) = x_{k}$$

The proposition is proved.

6.2.6 Lemma For $i, j \in \{1, 2, 3\}$ the Pauli matrices satisfy the following commutation relations:

$$[\sigma_i, \sigma_j] = 2i \sum_{k=1}^3 \varepsilon_{ijk} \sigma_k ,$$

where for $i, j, k \in \{1, 2, 3\}$ the Levi-Civita symbol ε_{ijk} is defined by

$$\varepsilon_{ijk} = \begin{cases} 1 & \text{if } (i, j, k) \text{ is an even permutation of } (1, 2, 3), \\ -1 & \text{if } (i, j, k) \text{ is an odd permutation of } (1, 2, 3), \text{ and} \\ 0 & else. \end{cases}$$

6.2.7 Remark Recall that a permutation of (1, 2, 3) is even if and only if it is cyclic.

Proof. The commutation relations follow immediately from equations (A.6.2.2) and (A.6.2.3) in the proof of the preceding proposition. \Box

6.2.8 Theorem The matrices

$$\tau_1 = \frac{1}{i}\sigma_1 = \begin{pmatrix} 0 & -i \\ -i & 0 \end{pmatrix}, \quad \tau_2 = \frac{1}{i}\sigma_2 = \begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix}, \quad \tau_3 = \frac{1}{i}\sigma_3 = \begin{pmatrix} -i & 0 \\ 0 & i \end{pmatrix}$$

form a basis of the Lie algebra $\mathfrak{su}(2)$ and obey the commutation relations

$$[\tau_i, \tau_j] = 2\tau_k$$
 for every cyclic permutation (i, j, k) of $(1, 2, 3)$. (A.6.2.4)

Moreover, the linear map $\Phi : \mathfrak{su}(2) \to \mathbb{R}^3$ uniquely defined by $\tau_k \mapsto 2e_k$ for k = 1, 2, 3 is an isomorphism of Lie algebras, where \mathbb{R}^3 carries the Lie algebra structure given by the cross product \times . In particular, the Lie algebras $\mathfrak{su}(2)$ and $\mathfrak{so}(3)$ are isomorphic with an isomorphism given by the composition

$$M \circ \Phi : \mathfrak{su}(2) \to \mathfrak{so}(3), \quad \sum_{k=1}^{3} x^{k} \tau_{k} = \begin{pmatrix} -x^{3}\mathbf{i} & -x^{2} - x^{1}\mathbf{i} \\ x^{2} - x^{1}\mathbf{i} & x^{3}\mathbf{i} \end{pmatrix} \mapsto 2 \begin{pmatrix} 0 & -x^{3} & x^{2} \\ x^{3} & 0 & -x^{1} \\ -x^{2} & x^{1} & 0 \end{pmatrix},$$

where $M : \mathbb{R}^3 \to \mathfrak{so}(3)$ is the isomorphism from Theorem 6.2.2.

Proof. Since multiplication by -i is a real linear isomorphism from $\mathfrak{Herm}^{tr0}(2)$ to $\mathfrak{su}(2)$ and since the Pauli matrices form a basis of $\mathfrak{Herm}^{tr0}(2)$, the matrices τ_k , k = 1, 2, 3, form a basis of $\mathfrak{su}(2)$. The commutation relations (A.6.2.4) are an immediate consequence of the preceding lemma. The Lie bracket is preserved by Φ since

 $(2e_i) \times (2e_j) = 2(2e_k)$ for every cyclic permutation (i, j, k) of (1, 2, 3).

The rest of the claim now follows by definition of Φ and Theorem 6.2.2.

6.2.9 Theorem For every $g \in SU(2)$ the linear map

$$\pi_g : \mathbb{R}^3 \to \mathbb{R}^3, \quad \vec{x} \mapsto \vec{\sigma}^{-1} \left(g(\vec{\sigma} \cdot \vec{x}) g^* \right)$$

is an orthogonal transformation. Moreover, the map

$$\pi: SU(2) \to SO(3), \quad g \mapsto \pi_q$$

is a differentiable surjective group homomorphism with kernel $\{\pm I_2\} \cong \mathbb{Z}/2$. In particular, π is the universal covering map of SO(3). Finally, the tangent map $T_1\pi : \mathfrak{su}(2) \to \mathfrak{so}(3)$ coincides with the isomorphism $M \circ \Phi$ from Theorem 6.2.8.

Proof. Observe that $det(gAg^*) = det A$ and $tr(gAg^*) = tr(A)$ for all $g \in SU(2)$ and $A \in \mathfrak{Herm}^{tr0}(2)$. This together with the fact that $\vec{\sigma}$ is an isometric isomorphism from $(\mathbb{R}^3, \|\cdot\|)$ to $(\mathfrak{Herm}^{tr0}(2), \sqrt{-det(\cdot)})$ entails that the transformations π_g are orthogonal.

A.6.3. The Lorentz group SO(1,3) and its universal cover $SL(2,\mathbb{C})$

to do: change signature from (-, +, +, +) back to (+, -, -, -).

6.3.1 Recall from Examples 6.1.3 (a) that the *Minowski inner product* of two elements $x = (x^0, x^1, x^2, x^3) \in \mathbb{R}^4$ and $y = (y^0, y^1, y^2, y^3) \in \mathbb{R}^4$ is defined by $\langle x, y \rangle_{\mathsf{M}} = x^0 y^0 - \sum_{k=1}^3 x^k y^k$, and that \mathbb{R}^4 endowed with the Minkowski inner product is denoted $\mathbb{R}^{1,3}$. The signature of the Minowski inner product therefore is (+, -, -, -) or in other terms (1, 3). As usual we call $\mathbb{R}^{1,3}$ *Minkowski space* of (*space-time*) dimension 4.

Recall from Examples 6.1.8 (b) that the *pseudo-orthogonal group* O(1,3) consists of all $g \in GL(4,\mathbb{R})$ such that

$$\langle gx, gy \rangle_{\mathsf{M}} = \langle x, y \rangle_{\mathsf{M}}$$
 for all $x, y \in \mathbb{R}^4$

Following common language in mathematical physics we call O(1,3) the *Lorentz group* in *space-time dimension* 4. The subgroup

$$SO(1,3) = \{g \in O(1,3) \mid \det g = 1\} \subset O(1,3)$$

is called the *proper Lorentz group*. Let us show that the Lorentz groups O(1,3) and SO(1,3) are Lie groups. To this end put

$$\eta = \begin{pmatrix} 1 & 0 & 0 & 0\\ 0 & -1 & 0 & 0\\ 0 & 0 & -1 & 0\\ 0 & 0 & 0 & -1 \end{pmatrix}$$

and observe that $\langle x, y \rangle_{\mathsf{M}} = \langle x, \eta y \rangle$ for all $x, y \in \mathbb{R}^4$, where $\langle -, - \rangle$ denotes the euclidean inner product. Hence a matrix $\Lambda \in \mathsf{GL}(4, \mathbb{R})$ lies in $\mathsf{O}(1, 3)$ if and only if

$$\Lambda^{t}\eta\Lambda - \eta = 0. \qquad (A.6.3.1)$$

Following standard language in mathematical physics we call every such Λ a *Lorentz transformation*. The map $f : GL(4, \mathbb{R}) \to \mathfrak{Sym}(4, \mathbb{R})$, $\Lambda \mapsto \Lambda^{t}\eta\Lambda - \eta$ is smooth and has derivative

$$T_{\Lambda}f: \mathfrak{Mat}(4,\mathbb{R}) \to \mathfrak{Sym}(4,\mathbb{R}), A \mapsto A^{\mathrm{t}}\eta\Lambda + \Lambda^{\mathrm{t}}\eta A$$

at $\Lambda \in GL(4, \mathbb{R})$. The derivative at Λ is surjective since $T_{\Lambda}f\left(\frac{1}{2}\eta(\Lambda^{t})^{-1}B\right) = B$ for all $B \in \mathfrak{Sym}(4, \mathbb{R})$. Hence f is a submersion and the preimage $O(1,3) = f^{-1}(0)$ a Lie subgroup of $GL(4, \mathbb{R})$. The Lie algebra $\mathfrak{o}(1,3)$ of the Lorentz group then consists of the kernel of $T_{\mathbb{I}}f$ that is of a all matrices $A \in \mathfrak{Mat}(4, \mathbb{R})$ such that

$$A^{t}\eta + \eta A = 0 . (A.6.3.2)$$

Since $\dim \mathfrak{o}(1,3) = \dim \mathfrak{Mat}(4,\mathbb{R}) - \dim \mathfrak{Sym}(4,\mathbb{R}) = 16 - 10 = 6$, one concludes that the Lorentz group O(1,3) is a Lie group of (real) dimension 6. By (A.6.3.1), the determinant of a Lorentz transformation $\Lambda \in O(1,3)$ fulfills $|\det(\Lambda)| = 1$. Moreover, time reversal

$$T = \begin{pmatrix} -1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \\ 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & 1 \end{pmatrix}$$

and parity inversion

$$P = \begin{pmatrix} 1 & 0 & 0 & 0\\ 0 & -1 & 0 & 0\\ 0 & 0 & -1 & 0\\ 0 & 0 & 0 & -1 \end{pmatrix}$$

are both Lorentz transformations with determinant -1. One concludes that SO(1,3) is a Lie subgroup of the Lorentz group O(1,3) and that the latter is the disjoint union of SO(1,3) and SO(1,3) $\cdot T = SO(1,3) \cdot P$.

6.3.2 The special linear group $SL(2, \mathbb{C})$ consists of all $g \in GL(2, \mathbb{C})$ such that $\det g = 1$. It is a complex Lie group by the following argument. Observe that the determinant $\det : GL(2, \mathbb{C}) \to \mathbb{C}$ is a complex differentiable group homomorphism. Its (complex) tangent map at the identity $\mathbb{1}$ is given by

$$T_{\mathbb{I}} \det : \mathfrak{gl}(2,\mathbb{C}) \to \mathbb{C}, \quad A \mapsto \left. \frac{\partial}{\partial z} \right|_{z=0} \det \exp(zA) = \left. \frac{\partial}{\partial z} \right|_{z=0} e^{z \operatorname{tr}(A)} = \operatorname{tr} A$$

This entails that $T_{\mathbb{1}} \det (z\mathbb{1}) = 2z$ for each $z \in \mathbb{C}$ hence det is a holomorphic submersion and $SL(2,\mathbb{C}) = \det^{-1}(1)$ a complex Lie group.

6.3.3 Proposition The Lie group $SL(2, \mathbb{C})$ is simply-connected.

Proof. We first show that SL(2) is path-connected. So let $g \in SL(2\mathbb{C})$. Then transform g into Jordan normal form that is choose $S \in GL(2,\mathbb{C})$ such that

$$SgS^{-1} = \begin{pmatrix} a_1 & e \\ 0 & a_2 \end{pmatrix} ,$$

where $a_1, a_2 \in \mathbb{C}$ with $a_1 a_2 = 1$ and $e \in \{0, 1\}$. Then choose a path $\gamma_1 : [0, 1] \to \mathbb{C}^{\times} = \mathbb{C} \setminus \{0\}$ such that $\gamma_1(0) = 1$ and $\gamma_1(1) = a_1$. Let $\gamma_2 : [0, 1] \to \mathbb{C}^{\times}$ be the path which maps t to $\gamma_2(t) = (\gamma_1(t))^{-1}$. Now put

$$h(t) = S^{-1} \begin{pmatrix} \gamma_1(t) & te \\ 0 & \gamma_2(t) \end{pmatrix}$$

Then $h : [0,1] \to SL(2,\mathbb{C})$ is a continuous path connecting $h(0) = I_2$ with h(1) = g. So $SL(2,\mathbb{C})$ is path-connected.

Next we prove that $SL(2, \mathbb{C})$ is simply-connected. To this end recall that the subgroup $SU(2) \subset$ $SL(2, \mathbb{C})$ is simply-connected. So to verify that $\pi_1(SL(2, \mathbb{C}))$ is trivial it suffices to construct a (strong) deformation retraction from $SL(2, \mathbb{C})$ onto SU(2) which means that we have to construct a continuous map $r : SL(2, \mathbb{C}) \times [0, 1] \rightarrow SL(2, \mathbb{C})$ such that

$$r_0 = \mathrm{id}, \quad r_1(\mathsf{SL}(2,\mathbb{C})) \subset \mathsf{SU}(2), \quad \text{and} \quad r_t|_{\mathsf{SU}(2)} = \mathrm{id}_{\mathsf{SU}(2)} \text{ for all } t \in [0,1] .$$

Here, as usual, r_t stands for the map $SL(2, \mathbb{C}) \to SL(2, \mathbb{C})$, $g \mapsto r(g, t)$.

Let us agree on the following notation. For every matrix $a = \begin{pmatrix} a_{11} & a_{12} \\ a_{21} & a_{22} \end{pmatrix} \in \mathfrak{Mat}(2 \times 2, \mathbb{C})$ we denote by a_i with i = 1, 2 the column vector $\begin{pmatrix} a_{1i} \\ a_{2i} \end{pmatrix}$ and write $a = (a_1, a_2)$. Vice versa, if $a_i = \begin{pmatrix} a_{1i} \\ a_{2i} \end{pmatrix} \in \mathbb{C}^2$ with i = 1, 2 are two (column) vectors then we denote by (a_1, a_2) the matrix $a = \begin{pmatrix} a_{11} & a_{12} \\ a_{21} & a_{22} \end{pmatrix}$. If now $g \in SL(2, \mathbb{C})$ with column vectors g_1, g_2 , then we know that g_1 and g_2 form a basis of \mathbb{C}^2 . Gram-Schmidt orthonormalization will transform the basis (g_1, g_2) into an orthonormal basis (u_1, u_2) :

$$(g_1, g_2) \mapsto (u_1, u_2) = \left(\frac{g_1}{\|g_1\|}, \frac{g_2 - \langle g_2, u_1 \rangle u_1}{\|g_2 - \langle g_2, u_1 \rangle u_1\|}\right)$$

Therefore, Gram–Schmidt orthonormalization can be understood as a retraction from $SL(2, \mathbb{C})$ to SU(2) leaving SU(2) invariant. So we are almost done, we just need to make the Gram–Schmidt process "continuous" in the sense that it can be deformed to the identity.

To achieve this define the following matrices depending on the parameter $t \in [0, 1]$:

$$p_t(g) = \begin{pmatrix} \frac{1}{\|g_1\|^t} & 0\\ 0 & 1 \end{pmatrix} , \quad q_t(g) = \begin{pmatrix} 1 & -t\langle g_2, u_1 \rangle \\ 0 & 1 \end{pmatrix} , \quad \widetilde{p}_t(g) = \begin{pmatrix} 1 & 0\\ 0 & \frac{1}{\|g_3\|^t} \end{pmatrix} ,$$

where $u_1 = \frac{g_1}{\|g_1\|}$ and $g_3 = g_2 - \langle g_2, u_1 \rangle u_1$. Each of these matrices lies in $GL(2, \mathbb{C})$ since their determinant is non-zero. Now we define $r : SL(2, \mathbb{C}) \times [0, 1] \rightarrow SL(2, \mathbb{C})$ by

$$r(g,t) = g \cdot p_t(g) \cdot q_t(g) \cdot \widetilde{p}_t(g)$$
, where $g \in SL(2,\mathbb{C}), t \in [0,1]$.

Then $r_0(g) = g$, $r_1(g) = (u_1, u_2) \in SU(2)$ (since (u_1, u_2) is an orthonormal basis of \mathbb{C}^2), r(g, t) = gif $g \in SU(2)$, and $r(g, t) \in SL(2, \mathbb{C})$ for all $g \in SL(2, \mathbb{C})$, $t \in [0, 1]$. The last property is the only not obvious one and needs to be verified because it guarantees that r is well-defined. The other properties are immediate and just tell that r is a strong deformation retraction of the kind we have been looking for.

We check two identies from which the remaining claim will follow immediately. For every $g = \begin{pmatrix} g_{11} & g_{12} \\ g_{21} & g_{22} \end{pmatrix} \in SL(2, \mathbb{C})$ compute

$$1 = \det g \cdot \overline{\det g} = (g_{11} g_{22} - g_{21} g_{12}) \cdot (\overline{g_{11} g_{22} - g_{21} g_{12}}) =$$

= $(|g_{11}|^2 |g_{22}|^2 + |g_{21}|^2 |g_{12}|^2) - 2\Re \mathfrak{e}(g_{11} g_{22} g_{21} g_{12}) =$
= $(|g_{11}|^2 |g_{12}|^2 + |g_{11}|^2 |g_{22}|^2 + |g_{21}|^2 |g_{12}|^2 + |g_{21}|^2 |g_{22}|^2) -$
 $- (|g_{11}|^2 |g_{12}|^2 + |g_{21}|^2 |g_{22}|^2 + 2\Re \mathfrak{e}(g_{11} g_{22} g_{21} g_{12})) = ||g_1||^2 ||g_2||^2 - |\langle g_1, g_2 \rangle|^2.$

Then

$$\det r(g,t) = \det g \det p_t(g) \det q_t(g) \det \widetilde{p}_t(g) = \left(\frac{1}{\|g_1\| \|g_3\|}\right)^t = \left(\frac{1}{\|g_1\|} \cdot \frac{\|g_1\|}{\sqrt{\|g_1\|^2 \|g_2\|^2 - |\langle g_1, g_2 \rangle|^2}}\right)^t = \left(\frac{1}{\sqrt{\|g_1\|^2 \|g_2\|^2 - |\langle g_1, g_2 \rangle|^2}}\right)^t = 1$$

which means that r(g,t) is in fact an element of $SL(2,\mathbb{C})$ for all $g \in SL(2,\mathbb{C})$ and $t \in [0,1]$. This finishes the proof.

6.3.4 Proposition Consider the space

$$\mathfrak{Herm}(2) = \left\{ X \in \mathfrak{gl}(2, \mathbb{C}) \mid X^* = X \right\}$$

of all hermitian 2×2 matrices. Then $\mathfrak{Herm}(2)$ is a real vector space of dimension 4 with a basis given by the identity matrix plus the Pauli matrices that is by

$$\sigma_0 = \begin{pmatrix} 1 & 0 \\ 0 & 1 \end{pmatrix}, \ \sigma_1 = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}, \ \sigma_2 = \begin{pmatrix} 0 & -i \\ i & 0 \end{pmatrix}, \ \sigma_3 = \begin{pmatrix} 1 & 0 \\ 0 & -1 \end{pmatrix}$$

The bilinear form

$$\langle \cdot, \cdot \rangle_{\mathfrak{h}} : \mathfrak{Herm}(2) \times \mathfrak{Herm}(2) \to \mathbb{R}, \quad (X, Y) \mapsto \frac{1}{2} \left(\det(X + Y) - \det X - \det Y \right)$$

is symmetric, non-degenerate, and has signature (1,3). An isometric isomorphism between $(\mathbb{R}^{1,3}, \langle \cdot, \cdot \rangle_{\mathsf{M}})$ and $(\mathfrak{Herm}(2), \langle \cdot, \cdot \rangle_{\mathfrak{h}})$ is given by

$$\sigma: \mathbb{R}^{1,3} \to \mathfrak{Herm}(2), \quad x \mapsto \sigma \cdot x = \sum_{k=0}^{3} x^k \sigma_k .$$

Its inverse maps $X \in \mathfrak{Herm}(2)$ to the vector x with components $x^k = \frac{1}{2} \operatorname{tr}(X\sigma_k)$, where k = 0, 1, 2, 3.

6.3.5 Theorem For every $g \in SL(2, \mathbb{C})$ the linear map

$$\pi_g : \mathbb{R}^4 \to \mathbb{R}^4, \quad x \mapsto \sigma^{-1} (g(\sigma \cdot x)g^*)$$

is a proper orthochronous Lorentz transformation. Moreover, the map

$$\pi: \mathsf{SL}(2,\mathbb{C}) \to \mathsf{SO}^{\uparrow}(1,3), \quad g \mapsto \pi_q$$

is a differentiable surjective group homomorphism with kernel $\{\pm I_2\} \cong \mathbb{Z}/2$. In particular, the tangent map $T_{\mathbb{I}}\pi : \mathfrak{sl}(2,\mathbb{C}) \to \mathfrak{so}(1,3)$ is an isomorphism and π is the universal covering map of $\mathsf{SO}^{\uparrow}(1,3)$.

Proof. Every element $g \in SL(2, \mathbb{C})$ induces a linear isomorphism

$$\alpha_g : \mathfrak{Herm}(2) \to \mathfrak{Herm}(2), \quad X \mapsto gXg^*$$

which is isometric since $det(\alpha_g X) = det X$ for all $X \in \mathfrak{Herm}(2)$. By Proposition 6.3.4, $\sigma : \mathbb{R}^{1,3} \to \mathfrak{Herm}(2)$ is an isometric isomorphism, hence $\pi_g = \sigma \circ \alpha_g \circ \sigma^{-1}$ leaves the Minkowski metric invariant and therefore is a Lorentz transformation.

A.7. Fiber bundles

A.7.1. Fiber bundles

Fibered manifolds and fibered charts

7.1.1 Definition By a *locpro-fibered manifold* we understand a smooth surjective submersion π : $E \to M$ from a locpro-manifold E onto a manifold M. If E is a finite dimensional manifold, one calls a surjective submersion $\pi : E \to M$ just a *fibered manifold*. One usually calls π the *projection*, E the *total space* and M the *base* of the (pro-)fibered manifold. A (locpro-) fibered manifold is often denoted as a triple (E, π, M) .

A morphism of (locpro-) fibered manifolds $\pi_1 : E_1 \to M_1$ and $\pi_2 : E_2 \to M_2$ consists of a pair (φ, f) of smooth maps $\varphi : E_1 \to E_2$ and $f : M_1 \to M_2$ such that the diagram

commutes. One sometimes also says in this situation that φ is a morphism of (locpro-) fibered manifolds over the map $f: M_1 \to M_2$. We in particular make use of this language when the base manifolds M_1 and M_2 coincide and f is the identity map. We then just say that φ is a morphism of (locpro-) fibered manifolds.

7.1.2 Obviously, the identity map id_E on the total space of a (locpro-) fibered manifold (E, π, M) is a morphism. Moreover, the composition

$$(\varphi_2, f_2) \circ (\varphi_1, f_1) := (\varphi_2 \circ \varphi_1, f_2 \circ f_1)$$

of morphisms

$$(\varphi_1, f_1) : (E_1, \pi_1, M_1) \to (E_2, \pi_2, M_2) \text{ and } (\varphi_2, f_2) : (E_2, \pi_2, M_2) \to (E_3, \pi_3, M_3)$$

is a morphism of (pro-)fibered manifolds from (E_1, π_1, M_1) to (E_2, π_2, M_2) , and id_E acts as identity morphism. One concludes that (locpro-)fibered manifolds and their morphisms form a category.

7.1.3 Definition Let (E, π, M) be a fibered manifold. By a *fibered chart* of (E, π, M) or a *chart* adapted to $\pi : E \to M$ one understands a chart $(U, \psi) :$.

7.1.4 Proposition Given a locpro-fibered manifold (E, π, M) , the fiber $F_p := \pi^{-1}(p)$ over an element $p \in M$ is a locpro-manifold.

Proof. In the case where E is finite dimensional the claim is an immediate consequence of the submersion theorem So assume that E is infinite dimensional. Since the claim is local, we can assume that there exists a smooth projective representation $(E_i, \eta_{ij}, \eta_i)_{i,j\in\mathbb{N}, i\leqslant j}$ of E. Since M is finite dimensional and using again that the claim is local we can assume that the the smooth map $\pi : E \to M$ factors in a neighborhood of F_p through some smooth map $\pi_i : E_i \to M$ that means that $\pi = \pi_i \circ \eta_i$. Since π is a smooth surjective submersion, π_i is so, too. Therefore, $F_i := \pi_i^{-1}(p)$ is a submanifold of E_i by the submersion theorem. Now put $\pi_j := \pi_i \circ \eta_{ij}$ for all j > i. As a composition of surjective submersions each such π_j is a surjective submersion as well. Hence for every j > i the preimage $F_j := \pi_j^{-1}(p) = \eta_{ij}^{-1}(F_i)$ for each $j \ge i$. Hence we obtain a smooth projective representation $(F_j, \varphi_{jk}, \varphi_j)_{j,k\in\mathbb{N}}$ of F_p , when defining φ_{jk} as the restriction $\eta_{jk}|_{F_k}$ and φ_j as the restriction $\eta_i|_{F_p}$. So F_p is a pro-manifold and the claim is proved.

7.1.5 Proposition A locpro-fibered manifold has local smooth sections that is for every locpro-fibered manifold (E, π, M) and every point $p \in M$ there exists as smooth map $s : U \to E$ defined on an open neighborhood U of p in M such that $\pi \circ s = id_U$.

A.8. Jets

A.8.1. A combinatorial interlude

Multi-Indices

8.1.1 Assume that \mathcal{I} is a non-empty set which we call *index set*. By a *multi-index* over \mathcal{I} we then understand an element $\alpha \in \mathbb{N}^{(\mathcal{I})}$ that is a family $\alpha = (\alpha_i)_{i \in \mathcal{I}}$ of natural numbers such that only finitely many α_i are non-zero. The *order* of such a multi-index is defined by $|\alpha| := \sum_{i \in \mathcal{I}} \alpha_i$. For $k \in \mathbb{N}$ and $k_1 \in \mathbb{N}$ and $k_2 \in \mathbb{N} \cup \{\infty\}$ with $k_1 \leq k_2$ we denote by $\mathbb{N}^{(\mathcal{I})}_k$ the set of all multi-indices over \mathcal{I} of order k and by $\mathbb{N}^{(\mathcal{I})}_{k_1,k_2}$ the set of all multi-indices of order less or equal k_2 which have order greater or equal k_1 . For reasons of clarity, which will be become obvious below, we sometimes also refer to an element of $\mathbb{N}^{(\mathcal{I})}$ as a *Greek multi-index*.

8.1.2 Example In most cases the index set \mathcal{I} will be of the form $\mathcal{I} = \{1, \ldots, d\}$ or of the form $\mathcal{I} = \{0, \ldots, d-1\}$ for some positive integer d. One then has $\mathbb{N}^{(\mathcal{I})} = \mathbb{N}^{\mathcal{I}} = \mathbb{N}^{d}$ and multi-indices are given by d-tuples of the form $\alpha = (\alpha_1, \ldots, \alpha_d)$ or $\beta = (\beta_0, \ldots, \beta_{d-1})$, respectively.

8.1.3 The space of multi-indices $\mathbb{N}^{(\mathcal{I})}$ carries the structure of a module over the semiring \mathbb{N} in the sense of Johnson & Manes (1970) that is $\mathbb{N}^{(\mathcal{I})}$ together with componentwise addition is an abelian monoid, componentwise multiplication with scalars is associative, 0 acts as zero map, 1 acts as identity, and the distributivity laws hold true. Moreover, $\mathbb{N}^{(\mathcal{I})}$ is free over the family of multi-indices $(1_i)_{i\in\mathcal{I}}$ defined by

 $1_i(j) := \begin{cases} 1 & \text{for } j = i, \\ 0 & \text{else.} \end{cases}$

8.1.4 By a *Roman multi-index* of a given order $k \in \mathbb{N}_{>0}$ over some index set \mathcal{I} we understand an element I of the cartesian product \mathcal{I}^k . For k = 0 we define \mathcal{I}^0 as the set {O}, where O is a fixed set not appearing as an element of \mathcal{I} . We call O the *Roman multi-index of order* 0 over \mathcal{I} . We sometimes write |I| for the order of a Roman multi-index. Note that we denote elements of \mathcal{I}^k by capital Roman letters I, J, ... and their components by their respective small Roman letters i_l , j_l , and so on.

For $k \ge 1$ the symmetric group S_k acts in a canoncial way on \mathcal{I}^k . We denote the orbit space of this action by $\overline{\mathcal{I}^k}$ and the orbit through a Roman multi-index $I \in \mathcal{I}^k$ by \overline{I} . In other words \overline{I} is the equivalence class of all Roman multi-indices obtained from I by permutation of its components. For k = 0 we identify $\overline{\mathcal{I}^0}$ with \mathcal{I}^0 and \overline{O} with O.

For Roman multi-indices $I = (i_1, \ldots, i_k) \in \mathcal{I}^k$ and $J = (j_1, \ldots, j_l) \in \mathcal{I}^l$ of positive order we denote by I + J the multi-index $(i_1, \ldots, i_k, j_1, \ldots, j_l) \in \mathcal{I}^{k+l}$. Obviously, the equivalence class $\overline{I + J}$ depends only on the equivalence classes \overline{I} and \overline{J} , hence the operation + descends to a map

$$+: \mathfrak{I}^k/S_k \times \mathfrak{I}^l/S_l \to \mathfrak{I}^{k+l}/S_{k+l}$$

It is straightforward to see that this operation is associative and commutative. Next define

$$I + O = O + I = I$$
 and $\overline{I} + \overline{O} = \overline{O} + \overline{I} = \overline{I}$

for all Roman multi-indices I and put $\overline{\mathcal{I}^{\bullet}} := \mathcal{I}^{\bullet}/\sim$, where $\mathcal{I}^{\bullet} := \bigsqcup_{k \in \mathbb{N}} \mathcal{I}^{k}$ and \sim is orbit equivalence which defines two Roman multi-indices $I \in \mathcal{I}^{k}$ and $J \in \mathcal{I}^{l}$ as equivalent if k = l and $\overline{I} = \overline{J}$. Then $\overline{\mathcal{I}^{\bullet}} = \bigsqcup_{k \in \mathbb{N}} \overline{\mathcal{I}^{k}}$. Moreover, $\overline{\mathcal{I}^{\bullet}}$ together with + as binary operation and O as zero element becomes an abelian monoid.

Note that every $i \in \mathcal{I}$ can be regarded as a Roman multi-index of order 1 and that $\overline{i} = i$, so we have the sums $I + i = (i_1, \ldots, i_k, i)$ and $\overline{I} + i = \overline{(i_1, \ldots, i_k, i)}$. Sometimes we also write Ii respectively $\overline{I}i$ for these sums.

8.1.5 Lemma Assume that \mathfrak{I} is totally ordered by some order relation \leq . Then every element of \mathfrak{I}^k of order $k \in \mathbb{N}_{>0}$ has a unique representative $I = (i_1, \ldots, i_k) \in \mathfrak{I}^k$ such that $i_1 \leq i_2 \leq \ldots \leq i_k$. We call such a representative an ordered Roman multi-index or an increasing representative.

Proof. One proves the claim by induction on the order k. For k = 1 the claim is obvious. Assume that it holds for some k and let \overline{J} be a Roman multi-index of order k + 1. Let j_m be the maximum of the components j_1, \ldots, j_{k+1} , and let $\sigma \in S_{k+1}$ be the permutation switching m and k + 1 and acting by identity on the rest. By hypothesis there exists a permutation $\tau \in S_k$ such that $j_{\sigma\tau(1)} \leq \ldots \leq j_{\sigma\tau(k)}$. Put $\tau(k+1) = k+1$. Then $\tau \in S_{k+1}$ and $I = (j_{\sigma\tau(1)}, \ldots, j_{\sigma\tau(k+1)})$ is a representative of \overline{J} with the desired properties. This finishes the inductive step and the claim is proved.

8.1.6 Proposition Let \mathfrak{I} be an index set with a total order \leq on it and $\kappa : \mathbb{N}^{(\mathfrak{I})} \to \mathfrak{I}^{\bullet}$ the map which maps the zero map $0_{\mathfrak{I}} : \mathfrak{I} \to \mathbb{N}$ to O and a Greek multi-index α of positive order to the Roman multi-index

$$\left(\underbrace{i_1,\ldots,i_1,i_2,\ldots,i_2}_{\alpha_{i_1} \text{ times } \alpha_{i_2} \text{ times } \alpha_{i_2} \text{ times } \alpha_{i_l} \text{ times } \right),$$

where $i_1 < \ldots < i_l$ are the (pairwise distinct and ordered) elements $i \in \mathcal{I}$ with non-vanishing component α_i . Then the induced map $\overline{\kappa} : \mathbb{N}^{(\mathcal{I})} \to \overline{\mathcal{I}^{\bullet}}$, $\alpha \mapsto \overline{\kappa(\alpha)}$ is an isomorphism of monoids and maps the space $\mathbb{N}^{(\mathcal{I})}_k$ of Greek multi-indices of order k onto $\overline{\mathcal{I}^k}$. Moreover, if \mathcal{I} is finite, then $\mathbb{N}^{(\mathcal{I})}_k$ and $\overline{\mathcal{I}^k}$ are finite as well and both have cardinality given by

$$|\mathbb{N}_{k}^{\mathbb{J}}| = |\overline{\mathcal{I}^{k}}| = \frac{1}{k!} \prod_{l=0}^{k-1} (|\mathcal{I}|+l)$$

Proof. First we need to show that $\overline{\kappa}$ is a bijection. To this end let us make our notation somewhat more precise and choose for each $\alpha \in \mathbb{N}^{(\mathcal{I})} \setminus \{0_{\mathcal{I}}\}$ the elements $i_{1}^{\alpha}, \ldots, i_{l_{\alpha}}^{\alpha} \in \mathcal{I}$ so that $i_{1}^{\alpha} < \ldots < i_{l_{\alpha}}^{\alpha}$, $\alpha_{i_{i}^{\alpha}} > 0$ for $j = 1, \ldots, l_{\alpha}$ and $\alpha_{i} = 0$ for all $i \in \mathcal{I} \setminus \{i_{1}^{\alpha}, \ldots, i_{l_{\alpha}}^{\alpha}\}$. Then

$$\kappa(\alpha) = \left(\underbrace{i_1^{\alpha}, \dots, i_1^{\alpha}, i_2^{\alpha}, \dots, i_2^{\alpha}, \dots, i_{l_{\alpha}}^{\alpha}, \dots, i_{l_{\alpha}}^{\alpha}}_{\alpha_{i_1^{\alpha}} \text{ times}}, \underbrace{\alpha_{i_2^{\alpha}} \text{ times}}_{\alpha_{i_2^{\alpha}} \text{ times}}, \underbrace{\alpha_{i_{\alpha}^{\alpha}} \text{ times}}_{\alpha_{i_{\alpha}^{\alpha}} \text{ times}}\right)$$

By construction, $\kappa(\alpha)$ is of increasing form. For each element $\overline{\mathbf{I}} \in \overline{\mathcal{I}^{\bullet}}$ let I be the representative of increasing form. Define $\lambda(\overline{\mathbf{I}}) \in \mathbb{N}^{(\mathcal{I})}$ as follows. If $\overline{\mathbf{I}} = \overline{\mathbf{O}}$, put $\lambda(\overline{\mathbf{I}}) = 0_{\mathcal{I}}$. If $\overline{\mathbf{I}} \neq \overline{\mathbf{O}}$, let $i_{1}^{\mathrm{I}} < \ldots < i_{l_{I}}^{\mathrm{I}}$ be the elements of \mathcal{I} which appear in I. Then, for each $j = 1, \ldots, l_{I}$ define $\alpha_{i_{j}}^{\mathrm{I}}$ to be the number of times the index i_{j}^{I} appears in I. For $i \in \mathcal{I}$ not appearing among the i_{j}^{I} put $\alpha_{i}^{\mathrm{I}} = 0$. Then define

$$\lambda(\bar{\mathbf{I}}) = \alpha^{\mathbf{I}} = \left(\alpha_i^{\mathbf{I}}\right)_{i \in \mathcal{I}}.$$

So we obtain a map $\lambda : \overline{\mathcal{I}^{\bullet}} \to \mathbb{N}^{(\mathcal{I})}$. For given $\mathbf{I} \neq \mathbf{O}$ one has by definition $l_{\alpha^{\mathrm{I}}} = l_{I}$ and $i_{1}^{\alpha^{\mathrm{I}}} = i_{1}^{\mathrm{I}}, \ldots, i_{l}^{\alpha^{\mathrm{I}}} = i_{l}^{\mathrm{I}}$ where $l = l_{\alpha^{\mathrm{I}}} = l_{I}$. Moreover, the index $i = i_{j}^{\alpha^{\mathrm{I}}}, j = 1, \ldots, l$ appears in $\overline{\kappa}(\lambda(\overline{\mathbf{I}}))$ exactly α_{i}^{I} times which coincides with the number i appears in I. Hence $\overline{\kappa}(\lambda(\overline{\mathbf{I}})) = \overline{\mathbf{I}}$. Now assume $\alpha \in \mathbb{N}^{\mathrm{I}} \setminus \{0_{\mathcal{I}}\}$ to be given and let $\mathbf{I} = \kappa(\alpha)$. Then $l_{I} = l_{\alpha}$ and $i_{1}^{\mathrm{I}} = i_{1}^{\alpha}, \ldots, i_{l}^{\mathrm{I}} = i_{l}^{\alpha}$ for $l = l_{I} = l_{\alpha}$. For each of the indices $i = i_{j}^{I}, j = 1, \ldots, l$, the *i*-th component of $\lambda(\overline{\kappa}(\alpha))$ coincides with α_{i} . Hence $\lambda(\overline{\kappa}(\alpha)) = \alpha$, which finishes the proof that $\overline{\kappa}$ is a bijection with inverse λ .

By construction of κ one has $|\kappa(\alpha)| = |\alpha|$ for all Greek multi-indices α which entails that for every $k \in \mathbb{N}$ the bijection κ maps $\mathbb{N}_{k}^{(\mathcal{J})}$ onto $\overline{\mathcal{I}^{k}}$.

Also by construction it is clear that $\kappa(\alpha + \beta) = \kappa(\alpha) + \kappa(\beta)$ for all $\alpha, \beta \in \mathbb{N}^{(\mathcal{I})}$ and that $\kappa(0_{\mathcal{I}}) = 0$. Hence κ is a morphism of monoids.

Now we will prove the formula for the cardinality of $\mathbb{N}_k^{\mathfrak{I}}$ by double induction on k and the cardinality of the index set \mathfrak{I} . Obviously $|\mathbb{N}_0^{\mathfrak{I}}| = 1$, so the claim holds for k = 0 and all finite index sets. Assume that it holds for some natural k and all finite index sets. Now let \mathfrak{I} be an index set of cardinality 1. Then $|\mathbb{N}_{k+1}^{\mathfrak{I}}| = 1$ since there is only one natural number with absolute value k + 1. Next assume that the claim holds for k + 1 and all index sets of cardinality less than d. Let \mathfrak{I} be an index set of cardinality d. Order the elements of \mathfrak{I} in some way so that $\mathfrak{I} = \{i_1, \ldots, i_d\}$ and $i_1 < \ldots < i_d$. The set $\mathbb{N}_{k+1}^{\mathfrak{I}}$ is then the disjoint union of the set of all $\alpha \in \mathbb{N}_{k+1}^{\mathfrak{I}}$ such that $\alpha_{i_d} = 0$ and the set of all $\alpha \in \mathbb{N}_{k+1}^{\mathfrak{I}}$ such that $\alpha_{i_d} \geq 1$. The first of these sets has cardinality

$$|\mathbb{N}_{k+1}^{\{i_1,\dots,i_{d-1}\}}| = \frac{1}{k+1!} \prod_{l=0}^k \left(d-1+l\right) \,,$$

the second has cardinality

$$|\mathbb{N}_{k}^{\mathbb{J}}| = \frac{1}{k!} \prod_{l=0}^{k-1} \left(d+l\right)$$

since the map

$$\left\{ \alpha \in \mathbb{N}_{k+1}^{\mathcal{I}} \mid \alpha_{i_d} \ge 1 \right\} \to \mathbb{N}_k^{\mathcal{I}} : \alpha \mapsto (\alpha_{i_1}, \dots, \alpha_{i_{d-1}}, \alpha_{i_d} - 1) \in \mathbb{N}_k^{\mathcal{I}}$$

is a bijection. Hence

$$|\mathbb{N}_{k+1}^{\mathcal{I}}| = \frac{1}{k+1!} \prod_{l=0}^{k} \left(d-1+l\right) + \frac{1}{k!} \prod_{l=0}^{k-1} \left(d+l\right) = \frac{1}{k+1!} \left(d-1+k+1\right) \prod_{l=0}^{k-1} \left(d+l\right) = \frac{1}{(k+1)!} \prod_{l=0}^{k} \left(d+l\right)$$

and the induction step is finished. The claim is proved.

8.1.7 By a *block* of a Roman multi-index I of positive order we mean a Roman multi-index of the form

$$\mathbf{I}_B = (i_{b_1}, \ldots, i_{b_{|B|}}) ,$$

where *B* is a subset of $\{1, \ldots, k\}$ and the $b_1, \ldots, b_{|B|} \in \{1, \ldots, k\}$ are the elements of *B* in increasing order. One can now decompose a multi-index I into blocks as follows. Let $\{B_1, \ldots, B_r\}$ be a partition of $\{1, \ldots, k\}$ which we assume to be lexicographically ordered that means that $b_{11} < b_{21} < \ldots < b_{r1}$, where $B_j = \{b_{j1}, \ldots, b_{j|B_j|}\}$ and $b_{jm} < b_{jn}$ for $j = 1, \ldots, r$ and $1 \le m < n \le |B_j|$. To express that $\{B_1, \ldots, B_r\}$ is a lexicographically ordered partition of $\{1, \ldots, k\}$ by r non-empty sets we write

$$B_1 \sqcup \ldots \sqcup B_r = \{1, \ldots, k\} \quad \& \quad \emptyset < B_1 < \ldots < B_r .$$

Now put $I_j := I_{B_j}$ for j = 1, ..., r. Then the Roman multi-indices I and $I_1 + ... + I_r$ are equivalent which can be interpreted as I being decomposed into the r blocks $I_1, ..., I_r$. More precisely, we call the r-tupel of pairs $((I_1, B_1), ..., (I_r, B_r))$ a *decomposition of* I *into* r *blocks* and denote the space of such decompositions by $Block^r(I)$. Note that the cardinality of $Block^r(I)$ coincides with the Sterling number of the second kind ${k \atop r}$ which gives the number of ways the set $\{1, ..., k\}$ can be partitioned into r subsets.

Multipowers and multiderivatives

8.1.8 Let M be a manifold and $x = (x^1, \ldots, x^d) : U \to \mathbb{R}^d$ a local coordinate system. Let $I \in \{1, \ldots, d\}^k$ be a Roman multi-index of positive order k. Then the product

$$x^{\mathbf{I}} := x^{i_1} \cdot \ldots \cdot x^{i_k} \tag{A.8.1.1}$$

and, for every $f \in \mathcal{C}^{\infty}(U)$, the higher derivative

$$\frac{\partial^{|\mathbf{I}|}f}{\partial x^{\mathbf{I}}} := \frac{\partial^k f}{\partial x^{i_1} \cdot \ldots \cdot \partial x^{i_k}} \tag{A.8.1.2}$$

are both invariant under permutations of the components of I, hence depend only on the equivalence class \overline{I} . We therefore sometimes write $x^{\overline{I}}$ for x^{I} and $\frac{\partial^{|\overline{I}|}f}{\partial x^{\overline{I}}}$ for $\frac{\partial^{|\overline{I}|}f}{\partial x^{\overline{I}}}$. In order 0 one puts $x^{\overline{O}} := x^{O} := 1$ and $\frac{\partial^{|\overline{O}|}f}{\partial x^{\overline{O}}} := \frac{\partial^{|O|}f}{\partial x^{O}} := f$. For a multi-index $\alpha \in \mathbb{N}^{d}$ one defines as usual

$$x^{\alpha} := (x^1)^{\alpha_1} \cdot \ldots \cdot (x^d)^{\alpha_d}$$

and

$$\frac{\partial^{|\alpha|}f}{\partial x^{\alpha}} := \frac{\partial^{|\alpha|}f}{(\partial x^1)^{\alpha_1} \cdot \ldots \cdot (\partial x^d)^{\alpha_d}} := \left(\frac{\partial}{\partial x^1}\right)^{\alpha_1} \cdot \ldots \cdot \left(\frac{\partial}{\partial x^d}\right)^{\alpha_d} f \; .$$

If now α and I are related by $\overline{I} = \kappa(\alpha)$, then

$$x^{\overline{1}} = x^{lpha}$$
 and $rac{\partial^{|\overline{1}|}f}{\partial x^{\overline{1}}} = rac{\partial^{|lpha|}f}{\partial x^{lpha}}$

by definition of κ and invariance of the product respectively the higher derivative under permutations of components of the multi-index.

8.1.9 Remark Occasionally we need multipowers and multiderivatives over more general index sets. So let \mathcal{I} be an arbitrary but still finite index set. Assume that the components of a coordinate system $x: U \to \mathbb{R}^{\mathcal{I}}$ are labelled x^i where i runs through the elements of \mathcal{I} . For $I \in \mathcal{I}^k$ equations (A.8.1.1) and (A.8.1.2) then can be used again to define multipowers x^{I} and multiderivatives $\frac{\partial^{|\mathrm{I}|} f}{\partial x^{\mathrm{I}}}$. Note that both objects are invariant under permutations of components of I, too, so the corresponding expressions where I is replaced by $\overline{\mathrm{I}}$ are also well-defined. Now let $\alpha = (\alpha_i)_{i\in\mathcal{I}} \in \mathbb{N}^{\mathcal{I}}$ be a Greek multi-index and $f \in \mathbb{C}^{\infty}(U)$. One then defines

$$x^{\alpha} := \prod_{i \in \mathcal{I}} (x^i)^{\alpha_i}$$

and

$$\frac{\partial^{|\alpha|} f}{\partial x^{\alpha}} := \prod_{i \in \mathbb{J}} \left(\frac{\partial}{\partial x^i} \right)^{\alpha_i} f \; .$$

One finally checks that when $\overline{\kappa}(\alpha) = \overline{I}$ the equalities $x^{\alpha} = x^{\overline{I}}$ and $\frac{\partial^{|\alpha|}f}{\partial x^{\alpha}} = \frac{\partial^{|\overline{I}|}f}{\partial x^{\overline{I}}}$ still hold true in this more general situation.

The formula of Faà-di-Bruno

8.1.10 Theorem (Combinatorial form of Faà-di-Bruno's formula) Let \mathcal{I} and \mathcal{J} denote finite index sets. Assume that M and N are smooth manifolds and that we are given smooth charts $x : U \hookrightarrow \mathbb{R}^{\mathcal{I}}$ and $y : V \hookrightarrow \mathbb{R}^{\mathcal{J}}$ over open domains $U \subset M$ and $V \subset N$. Assume further that $\varphi : U \to V$ is a smooth map. Denote by $\varphi_j : U \to \mathbb{R}$ for $j \in \mathcal{J}$ its components that means that $\varphi = (\varphi^j)_{j \in \mathcal{J}}$. Finally let $I \in \mathcal{I}^k$ be a Roman multi-index of positive order k. Then for every $f \in \mathbb{C}^{\infty}(V)$ the following equality holds true:

$$\frac{\partial^{|\mathbf{I}|}(f \circ \varphi)}{\partial x^{\mathbf{I}}} = \sum_{r=1}^{k} \sum_{\substack{\mathbf{J} \in \mathcal{J}^{r} \\ \mathbf{J} = (j_{1}, \dots, j_{r})}} \sum_{\substack{B_{1} \sqcup \dots \sqcup B_{r} = \{1, \dots, k\} \\ \emptyset < B_{1} < \dots < B_{r}}} \left(\frac{\partial^{|\mathbf{J}|}f}{\partial y^{\mathbf{J}}} \circ \varphi\right) \cdot \frac{\partial^{|\mathbf{I}_{B_{1}}|}\varphi^{j_{1}}}{\partial x^{\mathbf{I}_{B_{1}}}} \cdot \dots \cdot \frac{\partial^{|\mathbf{I}_{B_{r}}|}\varphi^{j_{r}}}{\partial x^{\mathbf{I}_{B_{r}}}} \cdot \dots \cdot \frac{\partial^{|\mathbf{I}_{B_{r}}|}\varphi^{j_{r}}}{\partial x^{\mathbf{I}_{B_{r}}}}} \cdot \dots \cdot \frac{\partial^{|\mathbf{I}_{B_{r}}|}\varphi^{j_{r}}}{\partial x^{\mathbf{I}_{B_{r}}}}} \cdot \dots \cdot \frac{\partial^{|\mathbf{I}_{B_{r}}|}\varphi^{j_{r}}}{\partial x^{\mathbf{I}_{B_{r}}}} \cdot \dots \cdot \frac{\partial^{|\mathbf{I}_{B_{r}}|}\varphi^{$$

Proof. We prove the claim by induction on the length of the multi-index I. Assume to be given a Roman multi-index $I \in \mathcal{I}^k$ of length k = |I| = 1. Then there exists a unique $i \in \mathcal{I}$ such that I = (i). By the chain rule one computes

$$\frac{\partial^{|\mathbf{I}|}(f \circ \varphi)}{\partial x^{\mathbf{I}}} = \frac{\partial(f \circ \varphi)}{\partial x^{i}} = \sum_{j \in \mathcal{J}} \left(\frac{\partial f}{\partial y^{j}} \circ \varphi \right) \cdot \frac{\partial \varphi^{j}}{\partial x^{i}} = \sum_{\substack{\mathbf{J} \in \mathcal{J}^{1} \\ \mathbf{J} = (j_{1})}} \sum_{\mathbf{B}_{1} = \{1\}} \left(\frac{\partial^{|\mathbf{J}|} f}{\partial y^{\mathbf{J}}} \circ \varphi \right) \cdot \frac{\partial^{|\mathbf{I}_{B_{1}}|} \varphi^{j_{1}}}{\partial x^{\mathbf{I}_{B_{1}}}} ,$$

hence the claim holds true for k = 1.

Now assume that for some $k \ge 1$ the claim holds for all Roman multi-indices of order $\le k$ over \mathcal{I} . Assume that $I = (i_1, \ldots, i_{k+1})$ is a Roman multi-index of order k + 1 over \mathcal{I} . Then $I = K + i_{k+1}$, where $K = (i_1, \ldots, i_k)$ is a Roman multi-index of order k. Using the induction hypothesis for K, the product and the chain rule one obtains

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$$\begin{split} \frac{\partial^{[l]}(f\circ\varphi)}{\partial x^{\mathrm{I}}} &= \frac{\partial}{\partial x^{i_{k+1}}} \frac{\partial^{[\mathrm{K}]}}{\partial x^{\mathrm{K}}} f\circ\varphi = \\ &= \frac{\partial}{\partial x^{i_{k+1}}} \sum_{r=1}^{k} \sum_{\substack{J\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k\}}} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{K}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{K}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{K}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{K}_{B_{r}}}} = \\ &= \frac{\partial}{\partial x^{i_{k+1}}} \sum_{r=1}^{k} \sum_{\substack{J\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k\}}} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{I}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{I}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{I}_{B_{r}}}} = \\ &= \sum_{r=1}^{k} \sum_{\substack{J\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{j_{r+1}\in\mathcal{J}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k\}}} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{I}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{I}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{I}_{B_{r}}}} = \\ &= \sum_{r=1}^{k} \sum_{\substack{I\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k\}}\\ \varphi^{J}=(j_{1},\ldots,j_{r})}} \left(\frac{\partial^{[J]}f}{\partial x^{\mathrm{I}_{B_{r}}}} \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{I}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{I}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{I}_{B_{r}}}} + \\ &+ \sum_{r=1}^{k} \sum_{\substack{I\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k\}\\\\ \emptyset < B_{1}=(k+1)}} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{I}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{I}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{I}_{B_{r}}}} + \\ &+ \sum_{r=1}^{k} \sum_{\substack{J\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k+1\}\\\\ \emptyset < B_{1}=(k+1)}} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{I}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{I}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{I}_{B_{r}}}} + \\ &+ \sum_{r=1}^{k} \sum_{\substack{J\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k+1\}\\\\ \emptyset < B_{1}=(k+1)}} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}}{\partial x^{\mathrm{I}_{B_{1}}}} \cdot \ldots \cdot \frac{\partial^{[\mathrm{I}_{B_{r}}]}\varphi^{j_{r}}}{\partial x^{\mathrm{I}_{B_{r}}}} = \\ &= \sum_{r=1}^{k+1} \sum_{\substack{J\in\mathcal{J}^{r} \\ J=(j_{1},\ldots,j_{r})}} \sum_{\substack{B_{1}\sqcup\ldots\sqcup B_{r}=\{1,\ldots,k+1\}\\\\ \emptyset < B_{1}\subseteq(k,\ldots,k_{r})} \otimes \varphi^{J} \otimes B_{1}\subseteq(k,\ldots,k_{r})} \left(\frac{\partial^{[J]}f}{\partial y^{\mathrm{J}}} \circ\varphi \right) \cdot \frac{\partial^{[\mathrm{I}_{B_{1}}]}\varphi^{j_{1}}$$

This concludes the induction step and the theorem is proved.

A.8.2. Jet bundles

8.2.1 Let us fix in this section a smooth finite dimensional fiber bundle $\pi^E : E \to M$. Denote by F its typical fiber and put $d = \dim M$, $n = \dim F$. The dimension of the total space E then is given by dim E = d + n. Note that for each point $p \in M$ the fiber $F_p = (\pi^E)^{-1}(p)$ is diffeomorphic to F.

Recall that $\Gamma^{\infty}(\pi^{E})$ stands for the *sheaf of smooth local sections* of π^{E} . Its space of sections over an open $U \subset M$ consists of all smooth $s: U \to E$ such that $\pi^E \circ s = \mathrm{id}_U$ and is denoted by $\Gamma^{\infty}(U, \pi^E)$. When writing $s \in \Gamma^{\infty}(\pi^{E})$ we mean that s is a smooth local section of E defined over some open subset $U = \operatorname{dom} s \subset M$. If $p \in X$ is a point, then $\Gamma^{\infty}(p, \pi^{E})$ denotes the space of local smooth sections about p that is the space of all smooth sections $s: U \to E$ defined on an open neighborhood $U \subset X$ of p. We will write \mathcal{U}_p° for the filter basis of all open neighborhoods of p and $\Gamma_p^{\infty}(\pi^E)$ for the stalk of $\Gamma^{\infty}(\pi^{E})$ at p which is defined as the colimit

$$\Gamma_p^{\infty}(\pi^E) = \operatorname{colim}_{U \in \mathfrak{U}_p^{\infty}} \Gamma^{\infty}(U, \pi^E) = \Gamma^{\infty}(p, \pi^E) / \sim_p .$$
(A.8.2.1)

Here we have made use of the fact that the colimit can be represented as the quotient of $\Gamma^{\infty}(p, \pi^E)$ by the equivalence relation \sim_p , where equivalence $s_1 \sim_p s_2$ of two smooth sections $s_1 : U_1 \to E$ and $s_2 : U_2 \to E$ over open neighborhoods of p is defined by the existance of an open neighborhood $U \subset U_1 \cap U_2$ of p such that $s_1|_U := s_2|_U$. The equivalence class of a section $s \in \Gamma^{\infty}(p, \pi^E)$ is denoted $[s]_p$ and is called the *germ* of s at p. So in other words, $\Gamma_p^{\infty}(\pi^E)$ is the space of all germs of smooth sections at p. To distinguish \sim_p from the later defined m-equivalence we call the relation \sim_p germ equivalence at p.

8.2.2 Definition Let $p \in M$ be a point in the base manifold M and $k \in \mathbb{N} \cup \{\infty\}$. Two local smooth sections $s_1 : U_1 \to E$ and $s_2 : U_2 \to E$ defined over open neighborhoods of p are said to be k-equivalent at p if $s_1(p) = s_2(p)$ and if for every fibered chart $(x, u) : W \to \mathbb{R}^d \times \mathbb{R}^n$ of π^E such that $p \in \pi(W)$, every index $b \in \{1, \ldots, n\}$ and all $\alpha \in \mathbb{N}^d$ with $|\alpha| \leq k$ the equality

$$\frac{\partial^{|\alpha|}(u^{\mathbf{b}} \circ s_1)}{\partial x^{\alpha}}(p) = \frac{\partial^{|\alpha|}(u^{\mathbf{b}} \circ s_2)}{\partial x^{\alpha}}(p) \tag{A.8.2.2}$$

holds true.

8.2.3 Proposition and Definition Let $p \in M$ be a point and $k \in \mathbb{N} \cup \{\infty\}$. Then k-equivalence at p is an equivalence relation on $\Gamma^{\infty}(p, \pi^{E})$. It will be denoted by the symbol $\sim_{k,p}$. The k-equivalence class of a smooth section $s : U \to E$ at p will be written $j_{p}^{k}(s)$. It is called the k-jet of s at p. The set of such k-jets at p coincides with the quotient space $J_{p}^{k}(\pi^{E}) = \Gamma^{\infty}(p, \pi^{E}) / \sim_{k,p}$. The union

$$\mathsf{J}^{k}(E) = \mathsf{J}^{k}(\pi^{E}) = \bigcup_{p \in M} \mathsf{J}^{k}_{p}(\pi^{E})$$

will be called the space of k-jets of sections of the bundle $\pi^E : E \to M$. Finally, there is a projection $\pi^k = \pi^{J^k(E)} : J^k(\pi^E) \to M$ which maps a jet $j_p^k(s)$, $s \in \Gamma^{\infty}(p, \pi^E)$ to its footpoint p.

Proof. The relation of k-equivalence at p is obviously reflexive and symmetric by definition. It is also transitive by transitivity of equality. Hence k-equivalence at p is an equivalence relation indeed. The claim is proved.

8.2.4 Lemma The following statements are equivalent for two sections $s_1, s_2 \in \Gamma^{\infty}(p, \pi^E)$ such that $s_1(p) = s_2(p)$:

- (1) The local sections s_1 and s_2 are k-equivalent at p.
- (2) If $(x, u) : W \to \mathbb{R}^d \times \mathbb{R}^n$ is a fibered chart of π^E such that $p \in \pi(W)$, then for all $b \in \{1, ..., n\}$ and $I \in \{1, ..., d\}^l$ with $1 \leq l \leq k$:

$$\frac{\partial^{|\mathbf{I}|}(u^{\mathbf{b}} \circ s_1)}{\partial x^{\mathbf{I}}}(p) = \frac{\partial^{|\mathbf{I}|}(u^{\mathbf{b}} \circ s_2)}{\partial x^{\mathbf{I}}}(p) .$$
(A.8.2.3)

- (3) There exists a fibered chart $(x, u) : W \to \mathbb{R}^d \times \mathbb{R}^n$ of π^E with $p \in \pi(W)$ such that (A.8.2.2) holds true.
- (4) There exists a fibered chart $(x, u) : W \to \mathbb{R}^d \times \mathbb{R}^n$ of π^E with $p \in \pi(W)$ such that (A.8.2.3) holds true.

Proof. The claim is an immediate consequence of the formula of Faà-di-Bruno.

8.2.5 Next we want to define a topology on the jet space $J^k(\pi^E)$ so that $\pi^k : J^k(\pi^E) \to M$ becomes a (topological) fiber bundle.

A.9. Geometric PDEs

A.9.1. Linear differential operators over commutative rings

9.1.1 In this section, A will always denote a commutative unital algebra over a field of characteristic zero \Bbbk . The identity element of A will be denoted by 1. Let M, N be two A-modules. An element $a \in A$ then acts in two natural ways on the space $\operatorname{Hom}_{\Bbbk}(M, N)$ of \Bbbk -linear maps from M to N, namely by

$$a_*: \operatorname{Hom}_{\Bbbk}(M, N) \to \operatorname{Hom}_{\Bbbk}(M, N), \ f \mapsto a_*f = af = (M \ni m \mapsto af(m) \in N)$$
(A.9.1.1)

and

$$a^*$$
: Hom_k $(M, N) \to$ Hom_k $(M, N), f \mapsto a^*f = fa = (M \ni m \mapsto f(am) \in N)$. (A.9.1.2)

9.1.2 Proposition and Definition The actions a_* and a^* define two A-module structures on $\operatorname{Hom}_{\Bbbk}(M, N)$ which are called the canonical left and the canonical right A-module structures, respectively. These module structures commute.

Proof. In the following let $a, b \in A$ and $f, g \in \operatorname{Hom}_{\Bbbk}(M, N)$. Then one computes for $m \in M$

$$\begin{aligned} ((a+b)_*f)(m) &= (a+b)(f(m)) = a(f(m)) + b(f(m)) \\ &= (a_*f)(m) + (b_*f)(m) = (a_*f + b_*f)(m) , \\ (a_*(f+g))(m) &= a(f(m) + g(m)) = af(m) + ag(m) = (a_*f + a_*g)(m) , \\ (a_*b_*f)(m) &= a(bf(m)) = (ab)(f(m)) = ((ab)_*f)(m) , \\ (1_*f)(m) &= 1 \cdot f(m) = f(m) , \end{aligned}$$

and

$$\begin{aligned} \big((a+b)^*f\big)(m) &= f\big((a+b)m\big) = f(am) + f(bm) \\ &= \big(a^*f\big)(m) + \big(b^*f\big)(m) = \big(a^*f + b^*f\big)(m) \ , \\ \big(a^*(f+g)\big)(m) &= f(am) + g(am) = a^*f(m) + a^*g(m) = \big(a^*f + a^*g\big)(m) \ , \\ \big(a^*b^*f\big)(m) &= \big(b^*f\big)(am) = f\big((b(am))\big)\big) = f\big((ab)m\big) = \big((ab)_*f\big)(m) \ , \\ \big(1^*F\big)(m) &= F(1 \cdot m) = F(m) \ . \end{aligned}$$

This proves the module properties. It remains to show that $a_*b^*f = b^*a_*f$. But that is clear since for all $m \in M$

$$(a_*b^*f)(m) = a((b^*f)(m)) = a(f(bm)) = (a_*f)(bm) = (b^*a_*f)(m)$$
.

9.1.3 Remark By the preceding proposition $\operatorname{Hom}_{\Bbbk}(M, N)$ becomes an A-bimodule which is not symmetric, in general, unless for example M = N = A. We regard $\operatorname{Hom}_{\Bbbk}(M, N)$ always as an object in the category of A-bimodules. When we want to consider only the canonical left or the canonical right A-module structure on the space of \Bbbk -linear maps from M to N we write ${}_{A}\operatorname{Hom}_{\Bbbk}(M, N)$ and $\operatorname{Hom}_{\Bbbk,A}(M, N)$, respectively, for the resulting objects in the category of A-modules.

9.1.4 Definition For every $a \in A$ denote by $ad_a : Hom_{\Bbbk}(M, N) \to Hom_{\Bbbk}(M, N)$ the k-linear map $a_* - a^*$ and call it the *adjoint action* of a.

9.1.5 Lemma Let M, N, P be A-modules. Then one has for all $f \in \text{Hom}_{\Bbbk}(M, N)$, $g \in \text{Hom}_{\Bbbk}(N, P)$ and all $a, b \in A$

$$ad_{ab} f = a_*(ad_b f) + b^*(ad_a f) = a^*(ad_b f) + b_*(ad_a f)$$
, (A.9.1.3)

$$\mathrm{ad}_a(g \circ f) = (\mathrm{ad}_a g) \circ f + g \circ (\mathrm{ad}_a f) . \tag{A.9.1.4}$$

Proof. Compute by observing that the left and right A-module structures commute:

$$\operatorname{ad}_{ab} f = (ab)_* f - (ab)^* f = a_*(b_* f - b^* f) + b^*(a_* f - a^* f) = a_*(\operatorname{ad}_b f) + b^*(\operatorname{ad}_a f)$$
.

By symmetry in a and b the first claimed equality follows. For the second observe that $(a^*g) \circ f = g \circ (a_*f)$ and compute

$$ad_a(g \circ f) = a_*(g \circ f) - a^*(g \circ f) = (a_*g - a^*g) \circ f + g \circ (a_*f - a^*f) = = (ad_a g) \circ f + g \circ (ad_a f) .$$

9.1.6 Definition For all A-modules M, N the space $Diff^0(M, N)$ of *linear differential operators* of order 0 from M to N is defined as the set of $D \in Hom_k(M, N)$ such that

$$\operatorname{ad}_a D = 0 \quad \text{for all } a \in A \ .$$

Recursively, one defines the space $\mathcal{D}iff^k(M, N)$ of *linear differential operators* of order $\leq k + 1$ from M to N as the set of all $D \in \operatorname{Hom}_{\mathbb{k}}(M, N)$ such that

$$\operatorname{ad}_a D \in \operatorname{Diff}^k(M, N)$$
 for all $a \in A$.

The space $\operatorname{Der}_{\Bbbk}(A, N)$ of *derivations* in N is defined as the set of all $D \in \operatorname{Hom}_{\Bbbk}(A, N)$ for which the Leibniz rule holds that is for which

$$D(ab) = aD(b) + bD(a) \quad \text{for all } a, b \in A \ .$$

9.1.7 Remark By definition, $\mathcal{D}iff^0(M, N)$ coincides with the space $\operatorname{Hom}_A(M, N)$ of A-module maps from M to N. By induction on k it becomes clear that $\mathcal{D}iff^k(M, N)$ can be equivalently described as the set of all $D \in \operatorname{Hom}_k(M, N)$ such that

$$(\operatorname{ad}_{a_0} \circ \ldots \circ \operatorname{ad}_{a_k}) D = 0 \quad \text{for all } a_0, \ldots, a_k \in A .$$

9.1.8 Proposition Let M, N, P be two A-modules. Then the following holds true for all $k, l \in \mathbb{N}$.

- (i) The space $\mathfrak{Diff}^k(M, N)$ inherits from $\operatorname{Hom}_{\Bbbk}(M, N)$ both A-module structures so is an A-subbimodule of $\operatorname{Hom}_{\Bbbk}(M, N)$. The two A-module structures coincide on $\mathfrak{Diff}^0(M, N)$ but in general not on spaces of differential operators of higher order.
- (ii) One has a canonical inclusion

$$\mathcal{D}iff^k(M,N) \subset \mathcal{D}iff^{k+1}(M,N)$$
.

- (iii) The composition of a differential operator $\Delta \in Diff^k(N, P)$ with a differential operator $D \in Diff^l(M, N)$ is a linear differential operator of degree $\leq k + l$.
- (iv) The space of derivations $\text{Der}_{\Bbbk}(A, N)$ is an A-submodule of $\mathcal{D}iff^{1}(M, N)$ with respect to the canonical left A-module structure but in general not an A-submodule of $\mathcal{D}iff^{1}(M, N)$ with respect to the canonical right A-module structure.

Proof. ad (i). The claim for $\mathcal{D}iff^0(M, N)$ holds since for every $D \in \mathcal{D}iff^0(M, N)$ and $a \in A$ the operators a_*D and a^*D coincide and are both A-linear again by the following equalities.

$$(a^*D)(m) = D(am) = D(am) = a(D(m)) = (a_*D)(m)$$
 for all $m \in M$ and
 $(a_*D)(bm) = a(D(bm)) = ab(D(m)) = b(aD(m)) = b(a_*D(m))$ for all $b \in A, m \in M$

Under the assumption that $\mathcal{D}iff^k(M, N)$ inherits the A-bimodule structure from $\operatorname{Hom}_{\Bbbk}(M, N)$ one checks for $D \in \mathcal{D}iff^{k+1}(M, N)$

$$ad_b(a_*D) = b_*a_*D - b^*a_*D = a_*(b_*D - b^*D) = a_*(ad_b D) \in Diff^k(M, N) \text{ and} \\ ad_b(a^*D) = b_*a^*D - b^*a^*D = a^*(b_*D - b^*D) = a^*(ad_b D) \in Diff^k(M, N) .$$

By induction $\mathcal{D}iff^k(M, N)$ therefore is an A-subbimodule of $\operatorname{Hom}_{\mathbb{k}}(M, N)$ for all $k \in \mathbb{N}$. Even though the two A-module structures coincide on $\mathcal{D}iff^0(M, N)$ they do not on spaces of differential operators of order 1 (and higher) as Example 9.1.9 below shows.

ad (ii). This is obvious by definition and an inductive argument.

ad (iii). If k+l=0 the claim is clear since then both Δ and D are A-linear, hence their composition is so, too. Assume that for some natural n the claim holds for all $k, l \in \mathbb{N}$ with $k+l \leq n$. Then assume k+l=n+1 and let $\Delta \in Diff^k(N, P)$ and $D \in Diff^l(M, N)$. Now compute using Equation (A.9.1.4)

$$\operatorname{ad}_a(\Delta \circ D) = (\operatorname{ad}_a \Delta) \circ D + \Delta \circ (\operatorname{ad}_a D)$$
.

By inductive hypothesis the right hand side is a differential operator of order $\leq n$, hence $\Delta \circ D \in Diff^{k+l}(M, P)$.

ad (iv). The space of derivations $\text{Der}_{\Bbbk}(A, N)$ is an A-submodule of $\mathcal{D}iff^{1}(M, N)$ with respect to the canonical left A-module structure. Namely, if $D \in \text{Der}_{\Bbbk}(A, N)$ and $a, b, c \in A$, then

$$(a_*D)(bc) = aD(bc) = abD(c) + acD(b) = b(aD(c)) + c(aD(b)) = b(a_*D)(c) + c(a_*D)(b) .$$

In general, $\text{Der}_{\Bbbk}(A, N)$ is not an A-submodule of $\mathcal{D}iff^1(M, N)$ with respect to the canonical right A-module structure.

9.1.9 Example Let $A = \Bbbk[X_1, \ldots, X_n]$ be the polynomial ring over \Bbbk in n indeterminates and $\Omega^1_{A/\Bbbk}$ the space of Kähler differentials of A that is the space I/I^2 , where I is the kernel of the multiplication map $\mu : A \otimes_{\Bbbk} A \to A$. The canonical map $d : A \to \Omega^1_{A/\Bbbk}$, $a \mapsto da = 1 \otimes a - a \otimes 1 + I^2$ then is a derivation and $\Omega^1_{A/\Bbbk}$ an A-module which is free over the elements dX_1, \ldots, dX_n . If now $a \in A \setminus \Bbbk$, then

$$a^*d(1) = da \neq 0 ,$$

so a^*d can not be a derivation. Note that a_*d is a derivation by Proposition 9.1.8 (iv).

9.1.10 By Proposition 9.1.8 one has a (filtered) diagram in the category of A-bimodules

$$\mathcal{D}iff^0(M,N) \hookrightarrow \mathcal{D}iff^1(M,N) \hookrightarrow \dots \hookrightarrow \mathcal{D}iff^k(M,N) \hookrightarrow .$$
(A.9.1.5)

Its colimit exists and coincides with the union of the $\mathcal{D}iff^k(M, N)$, $k \in \mathbb{N}$. We will denote it by $\mathcal{D}iff(M, N)$ and call it the A-bimodules of linear differential operators from M to N.

9.1.11 Remark In case we want to consider the spaces $\mathcal{D}iff^k(M, N)$ and $\mathcal{D}iff(M, N)$ with their canonical left A-module structure, only, we write ${}_A\mathcal{D}iff^k(M, N)$ and ${}_A\mathcal{D}iff(M, N)$, respectively. Analogously, when we regard $\mathcal{D}iff^k(M, N)$ and $\mathcal{D}iff(M, N)$ as objects in the category of A-modules with their canonical right A-module structure we denote them by $\mathcal{D}iff^k_A(M, N)$ and $\mathcal{D}iff_A(M, N)$, respectively. By $\mathcal{D}iff(M)$

9.1.12 Proposition Assigning to every pair of A-modules (M, N) the A-bimodule $\operatorname{Diff}^k(M, N)$ and to every pair of A-module maps $f: M' \to M$ and $g: N \to N'$ the A-bimodule map (f^*, g_*) : $\operatorname{Diff}^k(M, N) \to \operatorname{Diff}^k(M', N'), D \mapsto g \circ D \circ f$ comprises a bifunctor which is contravariant in the first and covariant in the second argument. Analogously, the assignment $(M, N) \to \operatorname{Diff}(M, N)$ becomes a bifunctor.

Proof. By definition, $((\mathrm{id}_M)^*, (id_N)_*)D = D$ for every $D \in \mathcal{D}iff^k(M, N)$, so

$$((\mathrm{id}_M)^*, (id_N)_*) = \mathrm{id}_{\mathrm{Diff}^k(M,N)}$$

Let $M_1, M_2, M_3, N_1, N_2, N_3$ denote A-modules and assume to be given A-modules maps $f_1: M_2 \rightarrow M_1, f_2: M_3 \rightarrow M_2, g_1: N_1 \rightarrow N_2$, and $g_2: N_2 \rightarrow N_3$. Then

$$\left(\left(f_2^*, g_{2*} \right) \circ \left(f_1^*, g_{1*} \right) \right) D = \left(f_2^*, g_{2*} \right) \left(g_1 \circ D \circ f_1 \right) = \left(g_2 \circ g_1 \right) \circ D \circ \left(f_1 \circ f_2 \right) = \\ = \left(\left(f_1 \circ f_2 \right)^*, \left(g_2 \circ g_1 \right)_* \right) D .$$

This proves that $Diff^k(-,-)$ and Diff(-,-) are bifunctors contravariant in the first and covariant in the second argument.

9.1.13 Theorem Let N be an A-module. Then the functors $Diff^k(-, N) : {}_A Mod \rightarrow {}_A Mod_A$ and $Diff(-, N) : {}_A Mod \rightarrow {}_A Mod_A$ are representable. Representing objects are given by the A-modules ${}_A Diff^k(N)$ and ${}_A Diff(N)$, respectively.

Back Matter

Bibliography

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Missing References

- MR [1] Section on manifolds-with-corners in the CRing Project
- MR [2] Section on examples of LF-spaces in the FANCy-Project

Theorem. Let $U \subset \mathbb{R}^n$ open and $\mathcal{D}(U) \subset \mathcal{C}^{\infty}(U)$ the space of test functions over U that is the space of all real valued smooth functions with support compact in U. Denote by $\mathcal{K}(U)$ the set of all compact subset $K \subset U$ and for all $K \in \mathcal{K}(U)$ by $\mathcal{D}_K(U)$ the space of real valued smooth functions with support in K. Endow $\mathcal{D}_K(U)$ with its natural structure of a Fréchet space. Finally, let $(K_k)_{k\in\mathbb{N}}$ be a compact exhaustion of U which means that each K_k has non-empty open interior, $\bigcup_{k\in\mathbb{N}} K_k = U$ and $K_k \Subset \mathring{K}_{k+1}$ for all $k \in \mathbb{N}$. Then the following locally convex structures on $\mathcal{D}(U)$ coincide:

- (i) the standard LF-space structure given by the locally convex colimit topology of the countable strict inductive system $(\mathcal{D}_{K_k}(U))_{k \in \mathbb{N}'}$
- (ii) the locally convex colimit topology of the inductive system $(\mathcal{D}_K(U))_{K\in\mathcal{K}(U)}$,
- (iii) the locally convex structure induced by the collection of all seminorms $q : \mathcal{D}(U) \to \mathbb{R}_{\geq 0}$ such that $q \circ \iota_K : \mathcal{D}_K(U) \to \mathbb{R}$ is a continuous seminorm, where $\iota_K : \mathcal{D}_K(U) \to \mathcal{D}(U)$ is the canoncial embedding,
- (iv) the locally convex structure induced by the collection of all seminorms $p_{N,\theta} : \mathcal{D}(U) \to \mathbb{R}_{\geq 0}$ of the form

$$p_{N,\theta}(f) = \sup_{\alpha \in \mathbb{N}^n, \, |\alpha| \leq N} \left\| \theta_{\alpha} \frac{\partial^{|\alpha|} f}{\partial^{\alpha} x} \right\|_{U} \quad \text{for } f \in \mathcal{D}(U) ,$$

where N runs through the elements of \mathbb{N} , θ through all locally finite families $(\theta_{\alpha})_{\alpha \in \mathbb{N}^n}$ of continuous functions $\theta_{\alpha} : U \to \mathbb{R}_{\geq 0}$, and where $\| - \|_U$ denotes the supremum norm over U.

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