D I P L O M A R B EIT

# Erkennung und 3D Posenschätzung Zusammengesetzter Objekte 

ausgeführt am<br>Institut für<br>Diskrete Mathematik und Geometrie<br>TU Wien<br>unter der Anleitung von

Associate Prof. Mag.rer.nat. Dr.techn. Christian Müller
durch
Florian Wimmer

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# DIPLOMATHESIS 

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supervised by
Associate Prof. Mag.rer.nat. Dr.techn. Christian Müller
by
Florian Wimmer


## Kurzfassung

Die orthogonale Gruppe $\mathrm{O}_{n}$ ist definiert als die Gruppe aller regulären $(n \times n)$-Matrizen $A$, deren transponierte Matrix $A^{T}$ die Inverse von $A$ ist. Die spezielle orthogonale Gruppe $\mathrm{SO}_{n}$ besteht aus allen orthogonalen $(n \times n)$-Matrizen mit Determinante 1 . Sie repräsentiert Rotationen um den Ursprung in $\mathbb{R}^{n}$. Die spezielle Euklidische Gruppe $\mathrm{SE}_{n}$ besteht aus allen Paaren ( $R, t$ ), wobei $R$ eine Rotation in $\mathrm{SO}_{n}$ und $t$ ein Vektor in $\mathbb{R}^{n}$ ist. Die Elemente von $\mathrm{SE}_{n}$ können die Posen von Objekten repräsentieren. Diese drei Untergruppen von $\mathrm{GL}_{n}$ sind differenzierbare Mannigfaltigkeiten.

Faktorgraphen sind bipartite Graphen mit Variablenknoten und Faktorknoten. Sie definieren die Faktorisierung einer Funktion und können die geometrischen Beziehungen verschiedener Objekte zueinander darstellen. Zusätzlich können Faktorgraphen eine probabilistische Struktur tragen.

Eine Retraktion ist eine Abbildung vom Tangentialbündel $T M$ einer glatten Mannigfaltigkeit $M$ auf $M$, die bestimmte Eigenschaften hat. Auf $\mathrm{SO}_{n}$ und $\mathrm{SE}_{n}$ können mithilfe der Exponentialfunktion für Matrizen Retraktionen definiert werden. Retraktionen ermöglichen die Anwendung iterativer Optimierungsmethoden auf Mannigfaltigkeiten analog zu Vektorräumen.

Im folgenden konkreten Anwendungsszenario werden Relativpositionen teilweise beweglicher Teile geschätzt. Betrachtet man einen Lastwagen als zusammengesetztes Objekt bestehend aus einfacheren Komponenten, wie zum Beispiel den Rädern des Lastwagens, erhält man eine Darstellung dieses zusammengesetzten Objekts als Faktorgraph. Die Variablenknoten des Faktorgraphen repräsentieren die verschiedenen Teile des Lastwagens, und die Faktorknoten die relativen Posen der Teile zueinander. Durch die Einführung eines Sensors, der einzelne Teile des Lastwagens beobachtet, erweitert sich dieser Faktorgraph. Für jeden Zeitschritt wird ein neuer Variablenknoten in den Faktorgraphen eingefügt, der den Sensor zu diesem Zeitpunkt repräsentiert. Die Beobachtungen des Sensors werden durch neue Faktorknoten dargestellt. Die Faktorknoten erhalten Wahrscheinlichkeitsdichten, wodurch die Berechnung einer maximalen a posteriori-Schätzung der Posen $X$ unter gegebenen Beobachtungen $Z$ möglich ist. Dabei wird die zusammengesetzte Wahrscheinlichkeitsfunktion $p(X, Z)$ mithilfe von Optimierung auf Mannigfaltigkeiten maximiert. Man erhält Schätzungen für die genaue Konfiguration des Lastwagens und die Pose des Sensors. Dieser Ansatz zur Posenschätzung zusammengesetzter Objekte kann mit dem Python-Paket GTSAM umgesetzt und getestet werden.


#### Abstract

The orthogonal group $\mathrm{O}_{n}$ is defined as the group of all invertible $(n \times n)$-matrices $A$ whose transposed matrix $A^{T}$ is the inverse of $A$. The special orthogonal group $\mathrm{SO}_{n}$ consists of all orthogonal $(n \times n)$-matrices with a determinant of 1 . It represents rotations around the origin in $\mathbb{R}^{n}$. The special Euclidean group $\mathrm{SE}_{n}$ comprises all pairs $(R, t)$, where $R$ is a rotation in $\mathrm{SO}_{n}$ and $t$ is a translation vector in $\mathbb{R}^{n}$. An element of $\mathrm{SE}_{n}$ can be used to represent the pose of an object. These three subgroups of $\mathrm{GL}_{n}$ are smooth manifolds.

Factor graphs are bipartite graphs with variable nodes and factor nodes and define the factorization of a function. They can encode geometrical relations among certain objects. Additionally, a factor graph can be equipped with a probabilistic structure.

A retraction is a mapping from the tangent bundle $T M$ of a smooth manifold $M$ to the manifold $M$ that satisfies certain properties, such as the local rigidity condition. By utilizing the exponential map for matrices, retractions can be defined on $\mathrm{SO}_{n}$ and $\mathrm{SE}_{n}$. Retractions allow simple implementations of iterative optimization techniques on manifolds.

In the following specific application scenario, the relative positions of partially movable components are estimated. Considering a truck as a composite object composed of simpler components, such as its wheels, leads to a representation of the truck as a factor graph. Variable nodes in the factor graph represent different parts of the truck, while factor nodes represent the relative poses of these parts to each other. Introducing a sensor observing specific parts of the truck expands the factor graph by adding variable nodes for the sensor at each time step and factor nodes for the observations. Equipping factor nodes with probability densities enables the computation of the maximum a posteriori estimate of some state $X$ given observations $Z$ by maximizing the joint probability function $p(X, Z)$ through optimization on manifolds. This approach provides estimates for the configuration of the truck and the pose of the sensor. Implementation and testing of this pose estimation method for composite objects can be achieved using the Python package GTSAM.


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## Eidesstattliche Erklärung

Ich erkläre an Eides statt, dass ich die vorliegende Diplomarbeit selbstständig und ohne fremde Hilfe verfasst, andere als die angegebenen Quellen und Hilfsmittel nicht benutzt bzw. die wörtlich oder sinngemäß entnommenen Stellen als solche kenntlich gemacht habe.

Wien, am 12.12.2023


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## 1. Introduction

Modern robot systems need robust and time-efficient techniques for detecting and locating objects in their environment. In this thesis, we discuss a novel technique for the estimation of the pose of composite objects, adapting existing methods used for related problems in robotics like Simultaneous Localization and Mapping (SLAM). This new technique includes the representation of composite objects as factor graphs and optimization on manifolds. In cooperation with the Austrian Institute of Technology (AIT), this approach was implemented and tested in an automated truck-loading scenario. Furthermore, we describe and implement an edge detection algorithm to detect the loading edge of a truck. Dellaert and Kaess present in [13] methods for modeling and solving problems in robotics with factor graphs. Here, we examine some mathematical background, especially the geometric aspects of the pose estimation problem.

The pose of an object in $\mathbb{R}^{3}$ is a distinguishable, static state of this object and can be represented by a matrix $T$ in the special Euclidean group $\mathrm{SE}_{3}$. It is commonly referred to as the position and orientation of this object. A matrix $T \in \mathrm{SE}_{3}$ has the form

$$
T=\left(\begin{array}{cc}
R & t \\
0 & 1
\end{array}\right) \in \mathbb{R}^{4 \times 4}
$$

for a vector $t \in \mathbb{R}^{3}$ indicating the position and a rotation matrix $R \in \mathrm{SO}_{3}$ indicating the orientation. The special orthogonal group $\mathrm{SO}_{3}$ is the subgroup of all matrices in $\mathrm{O}_{3}$ with determinant 1. The orthogonal group $\mathrm{O}_{3}$ is the group of all matrices $A \in \mathbb{R}^{3 \times 3}$ with

$$
A A^{T}=A^{T} A=I_{3}
$$

where $I_{3}$ denotes the identity matrix in $\mathbb{R}^{3 \times 3}$.
The estimation of the pose of an object can result in an optimization problem. Given a measurement $z \in \mathbb{R}^{n}$ and an estimation function $h: \mathrm{SE}_{3} \rightarrow \mathbb{R}^{n}$ that predicts measurements for given poses, we search for the matrix $T \in \mathrm{SE}_{3}$ that best approximates the measurements $z$ under the function $h$. Hence, we have to solve

$$
\underset{T \in \mathrm{SE}_{3}}{\arg \min }\|h(T)-z\|
$$

For this optimization problem, simple iterative optimization techniques like gradient descent fail. They rely on the updating rule

$$
x^{(t+1)}=x^{(t)}+\alpha \delta^{(t)}
$$

leading from the estimate $x^{(t)}$ in the time step $t$ to an improved estimate $x^{(t+1)}$ in the next time step by taking a step in the direction of $\delta^{(t)}$. The sum

$$
\left(\begin{array}{cc}
R_{1} & t_{1} \\
0 & 1
\end{array}\right)+\left(\begin{array}{cc}
R_{2} & t_{2} \\
0 & 1
\end{array}\right)=\left(\begin{array}{cc}
R_{1}+R_{2} & t_{1}+t_{2} \\
0 & 2
\end{array}\right)
$$

of two matrices $T_{1}$ and $T_{2}$ in $\mathrm{SE}_{3}$ is not in $\mathrm{SE}_{3}$ anymore. Even the sum $R_{1}+R_{2}$ of two rotation matrices $R_{1}, R_{2} \in \mathrm{SO}_{3}$ is in general not in $\mathrm{SO}_{3}$. Therefore, we cannot expect to receive a new valid estimate $T^{(t+1)} \in \mathrm{SE}_{3}$ by adding some matrix $\delta^{(t)}$ to $T^{(t)} \in \mathrm{SE}_{3}$. To work around this problem, we will exploit the structure of $\mathrm{SO}_{3}$ and $\mathrm{SE}_{3}$ as smooth manifolds and use retractions. With the exponential map

$$
e^{A}=\sum_{k \geq 0} \frac{A^{k}}{k!}
$$

for quadratic matrices $A$, we will define a retraction $\mathcal{R}$ that brings certain matrices back onto the manifolds $\mathrm{SO}_{3}$ respectively $\mathrm{SE}_{3}$.

To estimate the poses of different parts of a composite object simultaneously while considering the geometrical relations of the different parts to one another, we will represent the composite object as a factor graph. A factor graph is a bipartite graph with variable nodes and factor nodes that defines a factorization of a function. The variable nodes represent the different parts of the composite object and a factor node connected to two variable nodes represents the geometric transformation between the respective parts. The variable nodes define variables and the factor nodes are equipped with functions dependent on the variables of the nodes they are connected to. The functions of the factor nodes will define probability densities dependent on variables $X$ and observations $Z$. Consequently, the pose estimation problem for a composite object represented with a factor graph reads as a maximum a posteriori estimation problem of the joint probability function $p(X, Z)$, which can be reformulated as an optimization problem on the manifold $\mathrm{SE}_{3}$.

Dellaert and various contributors realized this factor graph and manifold optimization in the Python and C++ package GTSAM [11]. We use GTSAM for the implementation of the pose estimation of a truck viewed as a composite object.

This work consists of two main parts. In Chapter 2, we discuss the mathematical backgrounds for the implemented solutions of the loading edge detection and the pose estimation problem described in Chapter 3. The essential parts of the codes for our solutions to the loading edge detection problem and the pose estimation problem can be found in Appendix A and Appendix B.

## 2. Mathematical Foundations

On the surface, pose estimation as described and implemented in Section 3.2 looks like a software engineering problem. However, various mathematical concepts and considerations are necessary to enforce robust real-time pose estimation algorithms.
Here, the main mathematical concepts - either used implicitly as assumptions and foundations for programming and utilized Python packages, or explicitly as geometrical relations and algorithms in the implementation of this specific pose estimation problem - will be established and summarized.

First and foremost, we take in Section 2.1 a look at the geometric aspects of this problem. We define poses and pose spaces, mathematical groups and spaces related to this problem, and smooth manifolds. Throughout this Section, we investigate the rotation group $\mathrm{SO}_{3}$ from several perspectives.

Next, an important class of graphs will be introduced in Section 2.2. Factor graphs are the main idea of this pose estimation approach. Also, some common applications of factor graphs will be briefly described.

In Section 2.3, the employed optimization techniques are presented. On the one hand, we optimize on manifolds, in particular on $\mathrm{SO}_{3}$. This raises the issue of moving on the manifold to reach better solutions. On the other hand, we will see how to optimize a factor graph.

Finally, some standard algorithms and algorithmic concepts like RANSAC and point cloud manipulation as well as geometric data structures used in this project are described and analyzed in Section 2.4.

### 2.1. Geometry

The detection of objects in 3D space using traditional methods raises various geometric issues. It starts with the reconstruction of a 3D scene from multiple images taken by one or several cameras. Hartley and Zisserman describe in [29, Chapter 18] a few methods for solving these problems such as bundle adjustment. The data for the loading edge detection algorithm described in Section 3.1 and the pose estimation algorithm of a composite object described in Section 3.2 was obtained by using a special stereo camera which uses bundle adjustment to compute depth information.

Another issue is the processing of point clouds. Some aspects are covered in [16] and their realizations in the C++ and Python library Open3D are briefly described in [64]. In Section 3.1 we apply and discuss some point cloud processing techniques on the loading edge detection problem.

To detect an object, for example in a point cloud, it can be useful to view it in a simplified way as a geometric 3D shape that can easily be described mathematically. As a result, we
can exploit the well-known geometric properties in detection algorithms. For instance, the wheels of a truck as well as tree trunks resemble a right circular cylinder.

Going deeper into differential geometry, Dellaert and Kaess describe in [13, Chapter 6] methods for optimization on manifolds. Considering not only the position of an object in 3D space as a vector in $\mathbb{R}^{3}$, but also its orientation, raises the problem of how to search the space of possible solutions efficiently. The geometric foundation for this issue will be addressed in Section 2.1.2 and the actual optimization in Section 2.3.

These are just some of a variety of geometrical problems that arise in the surroundings of the tasks of object detection and pose estimation. In this Section, we start in 2.1.1 with the descriptions of mathematical groups like $\mathrm{O}_{n}, \mathrm{SO}_{n}$, and $\mathrm{SE}_{n}$. Different notions for describing the position and orientation of objects (especially in $\mathbb{R}^{3}$ ) are reviewed. Then in Section 2.1.2, we will dive into differential geometry to establish the basics for a geometric understanding and structure for $\mathrm{SO}_{n}$, discussed in 2.1.3. Furthermore, the exponential map for quadratic matrices is introduced and analyzed in 2.1.4, as we need it for optimization on $\mathrm{SO}_{n}$.

### 2.1.1. Matrix Groups and Poses

Representing an object's position and orientation is a crucial starting point for real-life geometric considerations [5]. The orthogonal group $\mathrm{O}_{n}$ and the special orthogonal group $\mathrm{SO}_{n}$ are matrix groups, studied in linear algebra. Here, they are investigated to define ways of denoting the orientation and therefore the pose of an object, especially in $\mathbb{R}^{3}$.

In the following, the definitions and properties of $\mathrm{O}_{n}$ and $\mathrm{SO}_{n}$ are based on [32, Chapter 12] and [26, Chapter 1]. In order to formally define this so-called pose of an object, some mathematical preparation is required. The general linear group $\mathrm{GL}_{n}(\mathbb{R})$, or from now on just $\mathrm{GL}_{n}$, is the group of all regular matrices in $\mathbb{R}^{n \times n}$ with the usual matrix multiplication as its group operation. Thus, these matrices represent all bijective linear maps from $\mathbb{R}^{n}$ to $\mathbb{R}^{n}$. Starting from $\mathrm{GL}_{n}$, we can define other matrix groups.

Definition 2.1.1. The orthogonal group $\mathrm{O}_{n}$ is the set of all matrices $A$ in $\mathbb{R}^{n \times n}$ that fulfill

$$
A A^{T}=A^{T} A=I .
$$

We have to check, whether the name group is justified for $\mathrm{O}_{n}$. In the following theorem, some basic properties of $\mathrm{O}_{n}$ are investigated.

Theorem 2.1.2 (Properties of $\mathrm{O}_{n}$ ). Let $n$ be a positive natural number. Then the following properties of the orthogonal group $\mathrm{O}_{n}$ hold.
(i) The orthogonal group $\mathrm{O}_{n}$ is a subgroup of $\mathrm{GL}_{n}$.
(ii) The column vectors of any matrix $A \in \mathrm{O}_{n}$ are pairwise orthogonal with respect to the Euclidean inner product $\langle\cdot, \cdot\rangle$ on $\mathbb{R}^{n}$.
(iii) The column vectors of any matrix $A \in \mathrm{O}_{n}$ have (Euclidean) norm 1 .
(iv) The column vectors of any matrix $A \in \mathrm{O}_{n}$ form an orthonormal basis of $\mathbb{R}^{n}$.
(v) For all $A \in \mathrm{O}_{n}$ the determinant $\operatorname{det} A$ is either +1 or -1 .
(vi) Any $A \in \mathrm{O}_{n}$ preserves the inner product on $\mathbb{R}^{n}$, i.e. $\langle x, y\rangle=\langle A x, A y\rangle$.
(vii) If $\lambda$ is an eigenvalue of an $A \in \mathrm{O}_{n}$, then $|\lambda|=1$ holds.

Proof. (i) First, since for all $A \in \mathrm{O}_{n}$ the property $A A^{T}=A^{T} A=I$ holds, the transposed matrix $A^{T}$ is the inverse of $A$. Therefore, $A$ is regular and $\mathrm{O}_{n} \subseteq \mathrm{GL}_{n}$.

For the identity matrix $I \in \mathbb{R}^{n \times n}$, we see that $I=I^{T}$ and $I I=I$. Thus, the multiplicative identity $I$ is in $\mathrm{O}_{n}$. Since $A^{T}=A^{-1}$ holds for all $A \in \mathrm{O}_{n}$, each element of $\mathrm{O}_{n}$ has its inverse element in $\mathrm{O}_{n}$. Furthermore, for any $A, B \in \mathrm{O}_{n}$ the computation

$$
(A B)(A B)^{T}=A B B^{T} A^{T} \stackrel{B \in \mathrm{O}_{n}}{=} A I A^{T} \stackrel{A \in \mathrm{O}_{n}}{=} I
$$

shows the closure of $\mathrm{O}_{n}$. Also, matrix multiplication is associative. Thus, $\mathrm{O}_{n}$ is a group and because of $\mathrm{O}_{n} \subseteq \mathrm{GL}_{n}$ a subgroup of the general linear group.
(ii) Let $A_{i}$ denote the $i$-th column vector of a matrix $A$ and therefore also the $i$-th row vector of the matrix $A^{T}$. The equation $I=A^{T} A$ for an $A \in \mathrm{O}_{n}$ translates to

$$
\left(\begin{array}{cccc}
1 & 0 & \cdots & 0  \tag{2.1}\\
0 & 1 & \cdots & 0 \\
\vdots & \vdots & \ddots & \vdots \\
0 & 0 & \cdots & 1
\end{array}\right)=I=A^{T} A=\left(\begin{array}{cccc}
\left\langle A_{1}, A_{1}\right\rangle & \left\langle A_{1}, A_{2}\right\rangle & \cdots & \left\langle A_{1}, A_{n}\right\rangle \\
\left\langle A_{2}, A_{1}\right\rangle & \left\langle A_{2}, A_{2}\right\rangle & \cdots & \left\langle A_{2}, A_{n}\right\rangle \\
\vdots & \vdots & \ddots & \vdots \\
\left\langle A_{n}, A_{1}\right\rangle & \left\langle A_{n}, A_{2}\right\rangle & \cdots & \left\langle A_{n}, A_{n}\right\rangle
\end{array}\right) .
$$

Thus, the product $\left\langle A_{i}, A_{j}\right\rangle$ for $i \neq j$ equals 0 , so the column vectors of $A$ are pairwise orthogonal.
(iii) From (2.1) directly follows $\left\langle A_{i}, A_{i}\right\rangle$ for any column $A_{i}$ of an orthogonal matrix $A$. Thus, the Euclidean norm of all columns of $A$ is equal to 1 .
(iv) From (i) we know that any $A \in \mathrm{O}_{n}$ is a regular matrix. Hence, the $n$ columns of an orthogonal matrix are linearly independent and therefore, they form a basis of $\mathbb{R}^{n}$. With (ii) and (iii), it follows that the columns of any $A \in \mathrm{O}_{n}$ form an orthonormal basis of $\mathbb{R}^{n}$.
(v) Since the determinant of a matrix is compatible with matrix multiplication and transposition, the equation $I=A A^{T}$ leads to

$$
1=\operatorname{det}(I)=\operatorname{det}\left(A A^{T}\right)=\operatorname{det}(A) \operatorname{det}\left(A^{T}\right)=\operatorname{det}(A)^{2}
$$

and thus $\operatorname{det}(A)= \pm 1$ for $A \in \mathrm{O}_{n}$.
(vi) The Euclidean inner product $\langle x, y\rangle$ can be viewed as $x^{T} y$ with $x$ and $y$ being column vectors in $\mathbb{R}^{n}$. Then we have $\langle A x, A y\rangle=(A x)^{T}(A y)=x^{T} A^{T} A y=x^{T} y=\langle x, y\rangle$.
(vii) The eigenvalues of $A \in \mathrm{O}_{n}$ are the solutions $\lambda$ of the equation $A v=\lambda v$. First, we take a look at the norm of the left side of the equation. In the following computation, we use the property $\langle x, y\rangle=\bar{x}^{T} y$ for the inner product in $\mathbb{C}^{n}$ (because eigenvectors of the real matrix $A$ can be in $\mathbb{C} \backslash \mathbb{R}$ ) and $A^{T}=A^{-1}$ and get

$$
\begin{aligned}
\|A v\|^{2} & =\langle A v, A v\rangle \\
& =\overline{(A v)}^{T}(A v) \\
& =\bar{v}^{T} A^{T} A v \\
& =\bar{v}^{T} v \\
& =\langle v, v\rangle=\|v\|^{2} .
\end{aligned}
$$

Thus, $\|v\|=\|A v\|=\|\lambda v\|=|\lambda|\|v\|$ implies $|\lambda|=1$. So we can conclude that all eigenvalues of an orthogonal matrix have an absolute value of 1 .

Remark 2.1.3. Property (vi) of the orthogonal group in Theorem 2.1.2 is widely used for an alternative, more general way of defining $\mathrm{O}_{n}$ on any vector space $V$ with an inner product $\langle\cdot, \cdot\rangle$ on $V$ : The orthogonal group $\mathrm{O}_{n}$ is the set of all automorphisms $f: V \rightarrow V$ that preserve the inner product, i.e. $\langle v, w\rangle=\langle f(v), f(w)\rangle[22, \S 0]$.

The automorphism $\mathbb{R}^{n} \rightarrow \mathbb{R}^{n}$ represented by a matrix $A \in \mathrm{O}_{n}$ is a reflection, a rotation, or a combination of reflection and rotation [26, Chapter 1]. As seen in Theorem 2.1.2, it preserves lengths and angles.

Example 2.1.4. The matrix

$$
\left(\begin{array}{ll}
0 & 1 \\
1 & 0
\end{array}\right) \in \mathrm{O}_{2}
$$

represents a reflection at the line with the equation $y=x$ in $\mathbb{R}^{2}$. The matrix

$$
\left(\begin{array}{rr}
0 & -1 \\
1 & 0
\end{array}\right) \in \mathrm{O}_{2}
$$

represents the rotation around the origin by an angle of $\frac{\pi}{2}$. So the matrix

$$
\left(\begin{array}{rr}
0 & -1 \\
1 & 0
\end{array}\right)\left(\begin{array}{ll}
0 & 1 \\
1 & 0
\end{array}\right)=\left(\begin{array}{rr}
-1 & 0 \\
0 & 1
\end{array}\right) \in \mathrm{O}_{2}
$$

represents the linear transformation that first reflects points at the line $y=x$ and then rotates them by $\frac{\pi}{2}$ around the origin.

Definition 2.1.5. The special orthogonal group $\mathrm{SO}_{n}$ is the set of all matrices $A \in \mathrm{O}_{n}$ with $\operatorname{det}(A)=1$.

Theorem 2.1.6. The special orthogonal group $\mathrm{SO}_{n}$ is a subgroup of $\mathrm{O}_{n}$.

Proof. The identity matrix $I \in \mathrm{O}_{n}$ has determinant 1 and is therefore in $\mathrm{SO}_{n}$. Since the inclusion $\mathrm{SO}_{n} \subseteq \mathrm{O}_{n}$ holds per definition, it holds that $A^{T}=A^{-1}$ for $A \in \mathrm{SO}_{n}$, and since $\operatorname{det}\left(A^{T}\right)=\operatorname{det}(A)=1$, the matrix $A$ has its inverse in $\mathrm{SO}_{n}$. To show the closure of $\mathrm{SO}_{n}$, we use the properties of determinants on $A, B \in \mathrm{SO}_{n}$ to get

$$
\operatorname{det}(A B)=\operatorname{det}(A) \operatorname{det}(B)=1 .
$$

Thus, $\mathrm{SO}_{n}$ is a group and because of $\mathrm{SO}_{n} \subseteq \mathrm{O}_{n}$ a subgroup of the orthogonal group.
Lemma 2.1.7. For an odd $n>1$, all $A \in \mathrm{SO}_{n}$ have 1 as an eigenvalue.
Proof. The eigenvalues of a matrix $A \in \mathrm{SO}_{n}$ are the zeros of the characteristic polynomial $\chi_{A}(\lambda)=\operatorname{det}(A-\lambda I)$. It holds with $A^{T}=A^{-1}, \operatorname{det}(A)=1$, and $\operatorname{det}(B)=\operatorname{det}\left(B^{T}\right)$ for all matrices $B \in \mathbb{R}^{n \times n}$

$$
\begin{aligned}
\operatorname{det}(A-I) & =\operatorname{det}\left(A-A A^{-1}\right) \\
& =\operatorname{det}\left(A\left(I-A^{-1}\right)\right) \\
& =\operatorname{det}(A) \operatorname{det}\left(I-A^{T}\right) \\
& =(-1)^{n} \operatorname{det}\left(A^{T}-I\right) \\
& =(-1)^{n} \operatorname{det}\left(\left(A^{T}-I\right)^{T}\right) \\
& =(-1)^{n} \operatorname{det}(A-I) .
\end{aligned}
$$

For an odd $n$, the equation reads as $\operatorname{det}(A-I)=-\operatorname{det}(A-I)$, so $\operatorname{det}(A-I)=0$ holds. Therefore, $\lambda=1$ is an eigenvalue of $A$.

For an even $n$, all eigenvalues have an absolute value of 1 (see Theorem 2.1.2(vii)). However, a polynomial with an even degree does not even have to have real roots as, for example, the characteristic polynomial of

$$
\left(\begin{array}{rr}
0 & -1 \\
1 & 0
\end{array}\right) \in \mathrm{SO}_{2}
$$

shows.
The special orthogonal group $\mathrm{SO}_{n}$ is also referred to as the rotation group. Geometrically, the matrices in $\mathrm{SO}_{2}$ represent planar rotations around the origin, and the matrices in $\mathrm{SO}_{3}$ represent rotations in $\mathbb{R}^{3}$ around an axis through the origin [26, Chapter 1].
Example 2.1.8 $\left(\mathrm{SO}_{2}\right)$. To get a matrix $A \in \mathrm{SO}_{2}$, the two column vectors $\left(a_{11}, a_{21}\right)^{T}$ and $\left(a_{12}, a_{22}\right)^{T}$ of $A$ have to be orthogonal, so

$$
\left\langle\binom{ a_{11}}{a_{21}},\binom{a_{12}}{a_{22}}\right\rangle=a_{11} a_{12}+a_{21} a_{22} \stackrel{!}{=} 0 .
$$

Since the column vectors have norm 1 , at least one of $a_{11}$ and $a_{21}$ has to be nonzero. Without loss of generality, let $a_{21} \neq 0$. The equation above translates to

$$
a_{22}=-\frac{a_{11}}{a_{21}} a_{12} .
$$

Therefore, the vector $\left(a_{12}, a_{22}\right)^{T}$ is uniquely defined by $\left(a_{11}, a_{21}\right)^{T}$ up to a scalar factor. Since both vectors have the same norm, we get

$$
\binom{a_{12}}{a_{22}}= \pm\binom{-a_{21}}{a_{11}} .
$$

To get unit length vectors, $a_{11}^{2}+a_{21}^{2}=1$ has to hold. So, $a_{11} \in[-1,1]$ follows. Let $\alpha$ be in $\left\{\arccos a_{11},-\arccos a_{11}\right\}$ to get $a_{11}=\cos \alpha$. With the well-known trigonometric property $\cos ^{2} \alpha+\sin ^{2} \alpha=1$ on the unit circle, we get $a_{21}=\sin \alpha$ if the sign of $\alpha$ was chosen accordingly. Furthermore, to get a positive determinant under these preconditions, we have to set $a_{22}=a_{11}$, because only then

$$
\operatorname{det}\left(\begin{array}{ll}
a_{11} & a_{12} \\
a_{21} & a_{22}
\end{array}\right)=a_{11} a_{22}-a_{21} a_{12}=\cos ^{2} \alpha-\sin \alpha(-\sin \alpha)=\cos ^{2} \alpha+\sin ^{2} \alpha=1
$$

holds.
We conclude that all matrices $A \in \mathrm{SO}_{2}$ are of the form

$$
A=\left(\begin{array}{rr}
\cos \alpha & -\sin \alpha \\
\sin \alpha & \cos \alpha
\end{array}\right)
$$

for an $\alpha \in[-\pi, \pi]$, which is the range of $\pm$ arccos on the real interval $[-1,1]$. Allowing only angles $\alpha \in(-\pi, \pi]$ gives a one-to-one correspondence between $\mathrm{SO}_{2}$ and the respective $\alpha$.

As seen in Example 2.1.8, any element of $\mathrm{SO}_{2}$ can be uniquely determined by one real number in $(-\pi, \pi]$, i.e. the angle of the corresponding planar rotation around the origin. We say that $\mathrm{SO}_{2}$ has one degree of freedom.
Lemma 2.1.9. Applying two rotations from $\mathrm{SO}_{2}$ by angles $\alpha$ and $\beta$ is the same as applying one rotation from $\mathrm{SO}_{2}$ by the angle $\alpha+\beta$.

Especially, the group $\mathrm{SO}_{2}$ with the usual matrix multiplication is commutative.
Proof. Let $A, B \in \mathrm{SO}_{2}$ be two matrices in the special orthogonal group. We have seen that there exist $\alpha, \beta \in(-\pi, \pi]$ such that

$$
A=\left(\begin{array}{rr}
\cos \alpha & -\sin \alpha \\
\sin \alpha & \cos \alpha
\end{array}\right) \quad \text { and } \quad B=\left(\begin{array}{rr}
\cos \beta & -\sin \beta \\
\sin \beta & \cos \beta
\end{array}\right) .
$$

Multiplying these two matrices yields

$$
\begin{aligned}
A B & =\left(\begin{array}{rr}
\cos \alpha & -\sin \alpha \\
\sin \alpha & \cos \alpha
\end{array}\right)\left(\begin{array}{rr}
\cos \beta & -\sin \beta \\
\sin \beta & \cos \beta
\end{array}\right) \\
& =\left(\begin{array}{cc}
\cos \alpha \cos \beta-\sin \alpha \sin \beta & -\cos \alpha \sin \beta-\sin \alpha \cos \beta \\
\sin \alpha \cos \beta+\cos \alpha \sin \beta & -\sin \alpha \sin \beta+\cos \alpha \cos \beta
\end{array}\right)
\end{aligned}
$$

Using sum identities for trigonometric functions, results in

$$
A B=\left(\begin{array}{rr}
\cos (\alpha+\beta) & -\sin (\alpha+\beta) \\
\sin (\alpha+\beta) & \cos (\alpha+\beta)
\end{array}\right)
$$

which is the matrix for a rotation by the angle $\alpha+\beta$.
Analogous computations for the matrix product $B A$ give the same result which shows commutativity.

The elements of $\mathrm{SO}_{3}$, the group of rotations in $\mathbb{R}^{3}$ [26, Chapter 1], do not behave as nicely as those of $\mathrm{SO}_{2}$. A rotation in $\mathbb{R}^{3}$ around an axis through the origin can be viewed as a planar rotation in the plane through the origin that is orthogonal to the rotation axis when we use an orthogonal projection to project $\mathbb{R}^{3}$ onto this plane. Hence, in the simple case of a rotation around a coordinate axis, we can use the representation of elements of $\mathrm{SO}_{2}$ we have derived above [32, p. 12.4.14]:

| rotation around the $x$-axis | rotation around the $y$-axis | rotation around the $z$-axis |
| :---: | :---: | :---: |
| $\left(\begin{array}{ccc}1 & 0 & 0 \\ 0 & \cos \alpha & -\sin \alpha \\ 0 & \sin \alpha & \cos \alpha\end{array}\right)$ | $\left(\begin{array}{ccc}\cos \alpha & 0 & -\sin \alpha \\ 0 & 1 & 0 \\ \sin \alpha & 0 & \cos \alpha\end{array}\right)$ | $\left(\begin{array}{ccc}\cos \alpha & -\sin \alpha & 0 \\ \sin \alpha & \cos \alpha & 0 \\ 0 & 0 & 1\end{array}\right)$ |

The angle $\alpha$ gives the rotation in the mathematically positive direction when viewed against the direction of the coordinate axis (e.g. rotations around the $z$-axis are viewed from the top). The coordinate corresponding to the rotation axis is fixed.

In general, multiplication in $\mathrm{SO}_{3}$ is not commutative, as a simple calculation with two rotation matrices around different axes shows.

To combine rotations with translations, we inspect two more groups.
Definition 2.1.10. The Euclidean group $\mathrm{E}_{n}$ is the set of all matrices of the form

$$
\left(\begin{array}{ll}
R & t \\
0 & 1
\end{array}\right)
$$

where $R$ is a matrix in $\mathrm{O}_{n}$ and $t$ is a translation vector in $\mathbb{R}^{n}$.
This set is also called the set of rigid transformations [20].
Note that the elements of $\mathrm{E}_{n}$ are in $\mathbb{R}^{(n+1) \times(n+1)}$. A translation $\mathbb{R}^{n} \rightarrow \mathbb{R}^{n}$ is not a linear transformation, thus it cannot be represented by an $(n \times n)$-matrix. Working with homogeneous coordinates $\left(x^{T}, 1\right)^{T}$ of a point $x \in \mathbb{R}^{n}$ instead, allows us to represent rotation and translation with one projective transformation [46].

Definition 2.1.11. The special Euclidean group $\mathrm{SE}_{n}$ is the set of all matrices of the form

$$
\left(\begin{array}{cc}
R & t  \tag{2.2}\\
0 & 1
\end{array}\right)
$$

where $R$ is a rotation matrix in $\mathrm{SO}_{n}$ and $t$ is a translation vector in $\mathbb{R}^{n}$.
This set is also called the set of proper rigid transformations [46].
Since $\mathrm{SO}_{n} \subseteq \mathrm{O}_{n}$, it holds that $\mathrm{SE}_{n} \subseteq \mathrm{E}_{n}$. Furthermore, the following theorem holds.
Theorem 2.1.12. The Euclidean group $\mathrm{E}_{n}$ and the special Euclidean group $\mathrm{SE}_{n}$ are subgroups of the general linear group $\mathrm{GL}_{n+1}$.

Proof. The proof for $\mathrm{E}_{n}$ and $\mathrm{SE}_{n}$ are completely analogous, so only the proof for $\mathrm{SE}_{n}$ is given.

First, we show that $\mathrm{SE}_{n}$ is a subset of $\mathrm{GL}_{n+1}$. The determinant of the block diagonal matrix $A \in \mathrm{SE}_{n}$ of the form (2.2) can be computed as

$$
\operatorname{det}\left(\begin{array}{ll}
R & t \\
0 & 1
\end{array}\right)=\operatorname{det}(R) \operatorname{det}(1)=1
$$

So, $A$ is a regular matrix with the shape $(n+1) \times(n+1)$ and is therefore contained in $\mathrm{GL}_{n+1}$.

We need to verify the group axioms for $\mathrm{SE}_{n}$ next. For $R=I_{n} \in \mathrm{SO}_{n}$ and $t$ being the zero vector in $\mathbb{R}^{n}$, the matrix of the form (2.2) is the identity element in $\mathrm{GL}_{n+1}$. Let

$$
A_{1}=\left(\begin{array}{cc}
R_{1} & t_{1} \\
0 & 1
\end{array}\right) \quad \text { and } \quad A_{2}=\left(\begin{array}{cc}
R_{2} & t_{2} \\
0 & 1
\end{array}\right)
$$

be elements of $\mathrm{SE}_{n}$. Then

$$
A_{1} A_{2}=\left(\begin{array}{cc}
R_{1} R_{2} & R_{1} t_{2}+t_{1} \\
0 & 1
\end{array}\right)
$$

which is in $\mathrm{SE}_{n}$, since the product $R_{1} R_{2}$ of two rotation matrices is again a rotation matrix and $R_{1} t_{2}+t_{1} \in \mathbb{R}^{n}$. If we want $A_{2}$ to be the inverse matrix of $A_{1}$, then $R_{1} R_{2}=I$ and $R_{1} t_{2}+t_{1}=0$ must both be true. So $R_{2}=R_{1}^{-1}$, which exists since rotation matrices are regular and their inverse is again a rotation matrix, and $t_{2}=-R_{1}^{-1} t_{1}$. To check if this is the inverse of $A_{1}$ indeed, we compute

$$
A_{2} A_{1}=\left(\begin{array}{cc}
R_{1}^{-1} & -R_{1}^{-1} t_{1} \\
0 & 1
\end{array}\right)\left(\begin{array}{cc}
R_{1} & t_{1} \\
0 & 1
\end{array}\right)=\left(\begin{array}{cc}
R_{1}^{-1} R_{1} & R_{1}^{-1} t_{1}-R_{1}^{-1} t_{1} \\
0 & 1
\end{array}\right)=I_{n+1} .
$$

As a result, $\mathrm{SE}_{n}$ is a group and altogether a subgroup of $\mathrm{GL}_{n+1}$.
We can uniquely identify the elements of $\mathrm{SE}_{n}$ with pairs in $\mathrm{SO}_{n} \times \mathbb{R}^{n}$ if needed:

$$
\left(\begin{array}{cc}
R & t  \tag{2.3}\\
0 & 1
\end{array}\right) \quad \longleftrightarrow \quad(R, t)
$$

On the one hand, a matrix $A \in \mathrm{SE}_{n}$ is the transformation matrix of a linear mapping from $\mathbb{R}^{n+1}$ to $\mathbb{R}^{n+1}$. On the other hand, the pair $T=(R, t) \in \mathrm{SE}_{n}$ can be interpreted as the function

$$
T:\left\{\begin{array}{l}
\mathbb{R}^{n} \rightarrow \mathbb{R}^{n}  \tag{2.4}\\
x \mapsto R x+t
\end{array} .\right.
$$

For an element of $\mathrm{SE}_{3}$, one can choose a rotation matrix $R \in \mathrm{SO}_{3}$ which has three degrees of freedom, and a translation vector $t \in \mathbb{R}^{3}$ where three values can be chosen independently. In total, $\mathrm{SE}_{3}$ has six degrees of freedom.

To illustrate $\mathrm{SO}_{3}$, we look at Euler's Rotation Theorem or as it is called in German Satz vom Fußball (theorem of the soccer ball) [18, Section 6.6].

Theorem 2.1.13 (Euler's Rotation Theorem / Satz vom Fußball). In a soccer match, if just one ball is used and it is placed at the beginning of the match and at the beginning of the second half of the match exactly at the kick-off point, then there are at least two points on the ball that are exactly in the same place both times.

Proof. During the first half of the match, the ball was subject to rotations and translations. Thus, the transformation of the ball can be represented as a product $T_{n} T_{n-1} \ldots T_{1}$ of transformations $T_{i} \in \mathrm{SE}_{3}$ for $i=1, \ldots, n$, where $T_{n}$ denotes the last transformation of the ball at the end of the first half back to the kick-off point. According to Theorem 2.1.12, $\mathrm{SE}_{3}$ is a group and the product is, therefore, some transformation $T=(R, t) \in \mathrm{SE}_{3}$. Since the ball is at the beginning of the second half at the same position as at the beginning of the first half, the translation part $t$ of $T$ is $(0,0,0)^{T} \in \mathbb{R}^{3}$. Thus, $T$ represents a rotation $R \in \mathrm{SO}_{3}$.

For simplicity, we assume the ball to be the unit sphere $S^{2} \subseteq \mathbb{R}^{3}$. To finish the proof, we have to find fixed points of $S^{2}$ under the rotation $R$, i.e. points $v \in \mathbb{R}^{3}$ with $R v=v$. Since $n=3$ is an odd number, Lemma 2.1.7 states that $R$ has an eigenvalue $\lambda=1$. This implies that there exists an (at least) one-dimensional subspace of $\mathbb{R}^{3}$ of eigenvectors $v$ that fulfill $R v=1 v$. A one-dimensional subspace of $\mathbb{R}^{3}$ is a line through the origin. The intersection of this line with $S^{2}$ is two (antipodal) points. These are the fixed points of the rotation $R$ which finishes the proof.

Remark 2.1.14. The statement of Theorem 2.1.13 can be rephrased as follows: Every matrix in $\mathrm{SO}_{3}$ represents a rotation in $\mathbb{R}^{3}$ around exactly one axis through the origin. Euler proved this version of the theorem in his paper [17] because he published the paper in 1775 and the game soccer in its now known form was invented in the $19^{\text {th }}$ century [62].

According to this theorem, a rotation in $\mathbb{R}^{3}$ can be defined by an axis through the origin and an angle $\alpha \in(-\pi, \pi]$. The axis can be defined by a point in $S^{2}$. So we can say that $\mathrm{SO}_{3}$ has three degrees of freedom [13, Chapter 6].

Now we try to define how to represent the position and orientation of an object, namely the pose of this object. Poses are frequently used in robotics and geometry but are rarely ever formally defined. In [13, Appendix B], a pose of a robot in $\mathbb{R}^{2}$ is defined as an element of $\mathrm{SE}_{2}$ with the respective rotation matrix in $\mathrm{SO}_{2}$ and a translation vector in $\mathbb{R}^{2}$. In [5, Section 2] Brégier et al. choose a more formal way and define poses generally for rigid objects. Our definition is based on [5, Section 2]. We will now focus on the typical use cases of poses, that is $\mathbb{R}^{2}$ or $\mathbb{R}^{3}$. So from now on in this section, $n \in\{2,3\}$ if not stated otherwise.

Definition 2.1.15. Given a rigid object in $\mathbb{R}^{n}$, a pose of this object is a distinguishable, static state of this object.

The pose space $\mathscr{P}$ of this object is the set of all possible poses $P$ of this object.
This definition of poses seems unnecessarily abstract for the naive intuition of the position and orientation of an object. This information could be described sufficiently by an element of $\mathrm{SE}_{2}$ or $\mathrm{SE}_{3}$. Definition 2.1.15 has an advantage over defining the pose of an object as an element of $\mathrm{SE}_{n}$ : Distinguishable in our definition means that an object in a pose $P_{1}$ can somehow be differentiated from the same object in a different pose $P_{2}$, which is a reasonable condition. Viewing the pose of an object with proper symmetry (e.g. the unit circle $S^{1} \subseteq \mathbb{R}^{2}$, a cube in $\mathbb{R}^{3}$, et cetera) as an element of $\mathrm{SE}_{n}$ would violate the condition of distinguishability since one pose can be represented by more than one element of $\mathrm{SE}_{n}$. Nevertheless, working with an element of $\mathrm{SE}_{n}$ seems to be easier than working with Definition 2.1.15. So we will identify the pose $P \in \mathscr{P}$ with an equivalence class of elements
of $\mathrm{SE}_{n}$ as in [5, Section 2.2]. For objects without proper symmetry these equivalence classes contain for all poses exactly one element of $\mathrm{SE}_{n}$ each. If an object is symmetric, the equivalence classes for the object's poses contain more than one element of $\mathrm{SE}_{n}$, depending on the symmetry class of the object. For deeper insights into symmetry classes of objects, we refer to the work of Schiller [52] that is related to this thesis.

We start with a reference pose $P_{0} \in \mathscr{P}$ for a rigid object in $\mathbb{R}^{n}$ that can be chosen arbitrarily. Given a new valid pose $P_{1} \in \mathscr{P}$ of this object, there exists a proper rigid transformation $T \in \mathrm{SE}_{n}$ as stated in (2.4) transforming the object from the pose $P_{0}$ to the pose $P_{1}$ in a way that every point $x$ of the object in the reference pose is transformed to a point $T(x)=R x+t$ of the object in the new pose $P_{1}$. We will also denote this as $T\left(P_{0}\right)=P_{1}$.

Definition 2.1.16. Given a rigid object in $\mathbb{R}^{n}$ for $n \in\{2,3\}$, a reference pose $P_{0}$ in the object's pose space $\mathscr{P}$, and two transformations $T, T^{\prime} \in \mathrm{SE}_{n}$, we say $T$ and $T^{\prime}$ generate the same pose if $T\left(P_{0}\right)=T^{\prime}\left(P_{0}\right) \in \mathscr{P}$. We write

$$
T \sim_{P_{0}} T^{\prime} \Leftrightarrow T\left(P_{0}\right)=T^{\prime}\left(P_{0}\right) .
$$

The relation $\sim_{P_{0}}$ is an equivalence relation since reflexivity, symmetry, and transitivity are directly derived from " $=$ ". Now we can properly identify a pose in the pose space with a set of proper rigid transformations.

Definition 2.1.17. Given a rigid object in $\mathbb{R}^{n}$ for $n \in\{2,3\}$ and a reference pose $P_{0}$ in the object's pose space $\mathscr{P}$, we define the class of proper rigid transformations representing a pose $P_{1} \in \mathscr{P}$ as

$$
\mathcal{T}_{P_{0}}^{P_{1}}:=\left\{T \in \mathrm{SE}_{n} \mid T\left(P_{0}\right)=P_{1}\right\} .
$$

The index $P_{0}$ can be omitted if it is clear or irrelevant which (fixed) pose is currently considered as the reference pose.

While a pose $P \in \mathscr{P}$ can refer to many elements of the special Euclidean group $\mathrm{SE}_{n}$, any element of $\mathrm{SE}_{n}$ belongs to exactly one pose $P \in \mathscr{P}$ and thus defines a pose of an object uniquely.

After defining poses and pose spaces formally, we return to the actual usage of poses. The translation part of a pose $P$ is relatively easy to handle, since for $(R, t) \in \mathcal{T}^{P}$ the translation vector $t$ is in $\mathbb{R}^{n}$, a well-known vector space with the Euclidean inner product. The rotation $R$ - until now viewed as an element of $\mathrm{SO}_{n}$ - is more challenging to manage. The mathematical structure of $\mathrm{SO}_{n}$ will be investigated in detail in Section 2.1.3. Dellaert and Kaess propose in [13, Appendix B] the most common types of representations of rotations in $\mathbb{R}^{2}$ and $\mathbb{R}^{3}$.

As stated above, an element of $\mathrm{SO}_{2}$ can be uniquely determined by a real number in the interval $(-\pi, \pi]$. Moreover, every real number can be interpreted as the angle of the rotation around the origin. By allowing every $\alpha \in \mathbb{R}$, we lose the uniqueness of identifying a number with a rotation, since rotating by the angle $\alpha$ is the same as rotating by the angle $\alpha+2 \pi$. There is a homomorphism between the groups $\mathbb{R}$ and $\mathrm{SO}_{2}$ given by the function that maps an $\alpha \in \mathbb{R}$ to the rotation in $\mathrm{SO}_{2}$ by the angle $\alpha$.

Another useful way to represent rotations in $\mathbb{R}^{2}$ are complex numbers. In the usual way, we identify $\mathbb{C}$ with $\mathbb{R}^{2}$ via

$$
z=\operatorname{Re} z+i \operatorname{Im} z \in \mathbb{C} \longleftrightarrow\binom{\operatorname{Re} z}{\operatorname{Im} z} \in \mathbb{R}^{2}
$$

Translating in $\mathbb{C}$ can be done by adding a number $z^{\prime} \in \mathbb{C}$. Rotating by an angle $\alpha$ can be achieved by multiplying with the complex number $\cos \alpha+i \sin \alpha$ which has length 1 . So the group of rotations is the set of complex numbers with length 1 together with multiplication in $\mathbb{C}$. This gives the one-to-one identification between rotations represented by the unit circle in $\mathbb{C}$ and rotations represented by $\mathrm{SO}_{2}$ :

$$
\cos \alpha+i \sin \alpha \longleftrightarrow\left(\begin{array}{rr}
\cos \alpha & -\sin \alpha \\
\sin \alpha & \cos \alpha
\end{array}\right)
$$

Rotations in $\mathbb{R}^{3}$ have various representations, too. The first way of representing a rotation is to describe it with an element of $\mathrm{SO}_{3}$ as depicted above. According to Theorem 2.1.13, each element of $\mathrm{SO}_{3}$ is a rotation around an axis through the origin. Therefore, we can represent a rotation as a pair $(a, \alpha) \in S^{2} \times \mathbb{R}$ of an axis $a$, given by a direction on the unit sphere $S^{2}$, and an angle $\alpha$. While easy to imagine, the description of a rotation in this way is not unique. For instance, the pairs $\left(\left(\frac{\sqrt{2}}{2}, \frac{1}{2},-\frac{1}{2}\right)^{T}, \frac{\pi}{3}\right)$ and $\left(\left(-\frac{\sqrt{2}}{2},-\frac{1}{2}, \frac{1}{2}\right)^{T},-\frac{\pi}{3}\right)$ describe the same rotation. Moreover, we know from Theorem 2.1.13 that some rotations around different axes with different angles yield again a rotation around an axis through the origin, but it is not as obvious as with matrices in $\mathrm{SO}_{3}$ how to combine the rotations.

The equivalent in $\mathbb{R}^{3}$ for complex numbers for 2D rotations are quaternions. Therefore, three pairwise different square roots $i, j, k$ of -1 are introduced. For more detailed information about quaternions, see, for example, [23].

Lastly, another intuitive way of representing rotations that outlines the three degrees of freedom of a 3D rotation is Euler angles. Often referred to as roll $\varphi$, pitch $\vartheta$, and yaw $\psi$, they compose a rotation as three consecutive rotations around different axes. There exist different conventions on the order of axes around which rotations occur. Sometimes the first rotation is around the $x$-axis, the second around the $y$-axis, and the third around the $z$-axis. (Proper) Euler angles are given for rotations around the $z$-, then the (rotated) $x$-, and then again the (rotated) $z$-axis. Also configurations like $x$, then $y$, then $x$ or $y$, then $z$, then $y$ are possible [24, Section 4.4].

From now on, we view the orientation of an object in $\mathbb{R}^{3}$ as a rotation matrix $R \in \mathrm{SO}_{3}$ if not stated otherwise.

### 2.1.2. Manifolds and Tangent Spaces

Pose estimation is the task of finding the pose that best fits some objective function with respect to certain preconditions and constraints [27]. As we know from above, the pose of an object represents its position and orientation. Just searching for the position that optimizes some (differentiable) function can be done with well-known methods such as gradient descent (see Section 2.3.1 for a revision of gradient descent). In general, we want to search in the neighborhood of a possible solution for a solution that is better with respect
to the objective function. On the one hand, getting a position in the neighborhood of the position of an object in $\mathbb{R}^{3}$ can be done by adding some small vector $v \in \mathbb{R}^{3}$. Searching for an orientation represented as a rotation matrix $R \in \mathrm{SO}_{3}$ that is somehow close to another orientation, on the other hand, cannot be done by adding an arbitrary, "small" matrix $V \in \mathbb{R}^{3 \times 3}$ to the matrix $R$. We know from Theorem 2.1.6 that $\mathrm{SO}_{n}$ with matrix multiplication is a group. But in general, $\mathrm{SO}_{n}$ is not closed with respect to matrix addition, as for instance the properties for matrices in $\mathrm{O}_{n}$ stated in Theorem 2.1.2 are not closed with respect to matrix or vector addition, so $R+V$ will not be in $\mathrm{SO}_{3}$, typically. Thus, we have to take a closer look at the geometrical structure of $\mathrm{SO}_{3}$ [13, Chapter 6]. The special orthogonal group $\mathrm{SO}_{n}$ is a manifold as we will see in 2.1.3. Here, we present the basics of manifolds and tangent spaces from differential geometry.

After introducing some elementary definitions, we will discuss the concept of smooth manifolds. First, we revise a few topological concepts.

Definition 2.1.18. (i) A Hausdorff space $(X, \mathcal{T})$ is a topological space that fulfills the $T_{2}$ separation axiom. Thus, for all points $x, y \in X$ with $x \neq y$ exist open neighborhoods $U_{x}$ of $x$ and $U_{y}$ of $y$ with $U_{x} \cap U_{y}=\emptyset[47$, Chapter $2 \S 17]$.
(ii) Let $X$ and $Y$ be topological spaces. A homeomorphism $f: X \rightarrow Y$ is a continuous, bijective function where the inverse function $f^{-1}$ is continuous as well [47, Chapter 2 §18].
(iii) An $n$-dimensional topological manifold $M$ is a topological space with a countable basis that is Hausdorff and has the property that for every point $x \in M$ there exists an open neighborhood $U_{x}$ of $x$ and an open set $V \subseteq \mathbb{R}^{n}$ such that $U_{x}$ is homeomorphic to $V$ [37, Section 2.2].
To properly define smooth manifolds on the basis of topological manifolds, we give some notions from differential geometry. The definitions are based on [37, Section 2.2].

Definition 2.1.19. Let $M$ be an $n$-dimensional topological manifold.
(i) Let $U \subseteq M$ be an open subset of $M$. For a homeomorphism $\varphi: U \rightarrow V$ into an open subset $V \subseteq \mathbb{R}^{n}$, the pair $(U, \varphi)$ is a chart of $M$.
For a point $x \in U$, we call $\varphi(x) \in \mathbb{R}^{n}$ the coordinates of $x$ in $(U, \varphi)$ [1, Section 3.1.1].
(ii) Let $\mathcal{A}=\left(\left(U_{i}, \varphi_{i}\right)\right)_{i \in I}$ be a family of charts of $M$ for some index set $I$. The family $\mathcal{A}$ is called an atlas of $M$ if $\bigcup_{i \in I} U_{i} \supseteq M$ holds.
(iii) The transition map between two charts $\left(U_{1}, \varphi_{1}\right)$ and $\left(U_{2}, \varphi_{2}\right)$ of $M$ with $U_{1} \cap U_{2} \neq \emptyset$ is the function

$$
\varphi_{2} \circ \varphi_{1}^{-1}: \varphi_{1}\left(U_{1} \cap U_{2}\right) \longrightarrow \varphi_{2}\left(U_{1} \cap U_{2}\right) .
$$

(iv) Two charts $\left(U_{1}, \varphi_{1}\right)$ and $\left(U_{2}, \varphi_{2}\right)$ are called $C^{k}$-compatible if their transition map is a $C^{k}$-diffeomorphism, i.e. $\varphi_{2} \circ \varphi_{1}^{-1}$ is bijective, $k$ times continuously differentiable and its inverse function $\left(\varphi_{2} \circ \varphi_{1}^{-1}\right)^{-1}$ is in $C^{k}$ as well.
If the transition map $\varphi_{2} \circ \varphi_{1}^{-1}$ is a $C^{\infty}$-diffeomorphism, we say the charts $\left(U_{1}, \varphi_{1}\right)$ and $\left(U_{2}, \varphi_{2}\right)$ are $C^{\infty}$-compatible or just compatible.
(v) A $C^{k}$-atlas $\mathcal{A}=\left(\left(U_{i}, \varphi_{i}\right)\right)_{i \in I}$ of $M$ is an atlas where the charts $\left(U_{i}, \varphi_{i}\right)$ and $\left(U_{j}, \varphi_{j}\right)$ are $C^{k}$-compatible for all $i, j \in I$ with $i \neq j$.
If the charts of $\mathcal{A}$ are pairwise compatible, we call $\mathcal{A}$ a $C^{\infty}$-atlas or smooth atlas.
(vi) A $C^{k}$ - or $C^{\infty}$-atlas $\mathcal{A}$ of a topological manifold $M$ is maximal if there is no chart $(U, \varphi)$ of $M$ that is $C^{k}$ - or $C^{\infty}$-compatible with all charts of $\mathcal{A}$ and not already contained in $\mathcal{A}$.

These definitions lead to smooth manifolds.
Definition 2.1.20. A $C^{k}$ - or $C^{\infty}$-manifold is a topological manifold provided with a maximal $C^{k}$ - or $C^{\infty}$-atlas.

We call a $C^{\infty}$-manifold also a smooth manifold.
Example 2.1.21. Let's take a look at the set $\mathbb{R}^{m \times n}$ of all $(m \times n)$-matrices with real entries for $m, n \in \mathbb{Z}^{+}$as in $\left[1\right.$, Section 3.1.5]. Let $\varphi: \mathbb{R}^{m \times n} \rightarrow \mathbb{R}^{m n}$ be the function defined by

$$
\varphi\left(\begin{array}{cccc}
a_{11} & a_{12} & \cdots & a_{1 n} \\
a_{21} & a_{22} & \cdots & a_{2 n} \\
\vdots & \vdots & \ddots & \vdots \\
a_{m 1} & a_{m 2} & \cdots & a_{m n}
\end{array}\right)=\left(\begin{array}{c}
a_{11} \\
\vdots \\
a_{m 1} \\
a_{12} \\
\vdots \\
a_{m n}
\end{array}\right)
$$

So $\varphi(A)$ is obtained by stacking the column vectors of the matrix $A$ on one another. This is a linear function between the two vector spaces $\mathbb{R}^{m \times n}$ and $\mathbb{R}^{m n}$. In the sense of Definition 2.1.18(ii), $\varphi$ is a homeomorphism since it is continuous, bijective and the inverse $\varphi^{-1}$ is bijective as well. The matrix space $\mathbb{R}^{m \times n}$ is an $m n$-dimensional topological manifold, because the topology, the Hausdorff property, and the second-countability are simply transferred via $\varphi^{-1}$ from $\mathbb{R}^{m n}$, and any open neighborhood $U_{A}$ of a point $A \in \mathbb{R}^{m \times n}$ is homeomorphic to the open set $\varphi\left(U_{A}\right)$ of $\mathbb{R}^{m n}$. Furthermore, $\left(\mathbb{R}^{m \times n}, \varphi\right)$ is a chart of $\mathbb{R}^{m \times n}$, and since it covers the whole space, we already have an atlas $\mathcal{A}$. We can add all charts $(U, \psi)$ to the atlas $\mathcal{A}$ that are compatible with the chart $\left(\mathbb{R}^{m \times n}, \varphi\right)$. This gives us a maximal smooth atlas. Thus, $\mathbb{R}^{m \times n}$ with this structure is a (smooth) manifold.

Given two manifolds $M_{1}$ and $M_{2}$ of dimensions $d_{1}$ and $d_{2}$, the product space $M_{1} \times M_{2}$ can be equipped with the product topology. For charts $\left(U_{1}, \varphi_{1}\right)$ and $\left(U_{2}, \varphi_{2}\right)$ of $M_{1}$ and $M_{2}$, the function

$$
\psi: U_{1} \times U_{2} \rightarrow \mathbb{R}^{d_{1}} \times \mathbb{R}^{d_{2}}:\left(x_{1}, x_{2}\right) \mapsto\left(\varphi_{1}\left(x_{1}\right), \varphi_{2}\left(x_{2}\right)\right)
$$

is a chart of $M_{1} \times M_{2}$ since all properties directly transfer from $\varphi_{1}$ and $\varphi_{2}$. Thus, two atlases $\mathcal{A}_{1}$ and $\mathcal{A}_{2}$ of $M_{1}$ and $M_{2}$ create an atlas $\mathcal{A}$ of $M_{1} \times M_{2}$. So, the product $M_{1} \times M_{2}$ is a manifold [1, Section 3.1.6].

In Section 2.3.2, we will make use of the tangent space $T_{x} M$ of a point $x$ in a manifold $M$ to optimize a function on the manifold $M$. There are various equivalent ways to define tangent vectors and spaces [40, Section 3.6], here we use smooth curves on the manifold as in [37, Section 2.6]. A smooth curve is a smooth function $\gamma: I \rightarrow M$ for an interval $I \subseteq \mathbb{R}$ (see [37, Chapter 2] for more about smooth functions on manifolds).

Definition 2.1.22. Let $M$ be an $n$-dimensional smooth manifold, $x \in M$ a point on $M$, and $\mathcal{C}_{x}^{M}$ the set of all smooth curves $\gamma: I \rightarrow M$ on $M$ with $0 \in I$ and $\gamma(0)=x$.
(i) Let $\gamma_{1}, \gamma_{2} \in \mathcal{C}_{x}^{M}$ be two curves through $x \in M$. They are called tangent at $x$ if there exists a chart $(U, \varphi)$ of $M$ with $x \in U$ and $\left(\varphi \circ \gamma_{1}\right)^{\prime}(0)=\left(\varphi \circ \gamma_{2}\right)^{\prime}(0)$. The function $\varphi \circ \gamma_{i}$ is a mapping $I \subseteq \mathbb{R} \rightarrow \mathbb{R}^{n}$, thus $\left(\varphi \circ \gamma_{i}\right)^{\prime}(0)$ is the ordinary derivative. This definition does not depend on the choice of the chart $(U, \varphi)$ (this can be verified with the chain rule of differentiation for another chart $(\tilde{U}, \tilde{\varphi})$ as in [37, Section 2.6]). Therefore, being tangent at a point $x$ defines an equivalence relation on $\mathcal{C}_{x}^{M}$.
(ii) A tangent vector to $M$ at $x$ is an equivalence class of the relation defined in (i).
(iii) The set of all tangent vectors to $M$ at $x$ is called the tangent space $T_{x} M$ to $M$ at $x$.
(iv) The tangent bundle

$$
T M:=\left\{(x, \xi) \mid x \in M, \xi \in T_{x} M\right\}
$$

of $M$ is the disjoint union of all tangent spaces to $M$.
To establish a vector space structure on a tangent space $T_{x} M$, we define a function $\vartheta_{\varphi}^{x, M}$ between the tangent space and $\mathbb{R}^{n}$ similar to [ 37 , Section 2.6.1].

Lemma 2.1.23. Let $M$ be an n-dimensional smooth manifold, $x$ a point on $M$, and $(U, \varphi)$ a chart of $M$ with $x \in U$. The function $\vartheta_{\varphi}^{x, M}$ defined as

$$
\vartheta_{\varphi}^{x, M}: T_{x} M \rightarrow \mathbb{R}^{n}: \xi=[\gamma] \mapsto(\varphi \circ \gamma)^{\prime}(0)
$$

is a bijection.
Proof. First, we have to show that $\vartheta_{\varphi}^{x, M}$ is well-defined. According to Definition 2.1.22(i) of the equivalence relation on the set $\mathcal{C}_{x}^{M}$ of smooth curves, a different representative $\tilde{\gamma} \in \xi$ gives the same value $(\varphi \circ \tilde{\gamma})^{\prime}(0)$ as $(\varphi \circ \gamma)^{\prime}(0)$. Thus, $\vartheta_{\varphi}^{x, M}(\xi)$ is independent of the choice of a curve $\gamma$ of $\xi$. Furthermore, $\varphi \circ \gamma$ has the codomain $\mathbb{R}^{n}$. Therefore, $\vartheta_{\varphi}^{x, M}$ is well-defined.

For two tangent vectors $\xi_{1}, \xi_{2} \in T_{x} M$ with representatives $\gamma_{i} \in \xi_{i}$ for $i \in\{1,2\}$ and $\xi_{1} \neq \xi_{2}$, it holds that $\left(\varphi \circ \gamma_{1}\right)^{\prime}(0) \neq\left(\varphi \circ \gamma_{2}\right)^{\prime}(0)$ since $\gamma_{1}$ and $\gamma_{2}$ would have been in the same equivalence class otherwise. Thus, the function $\vartheta_{\varphi}^{x, M}$ is injective.

Let $p \in \mathbb{R}^{n}$ be an arbitrary element of the codomain of $\vartheta_{\varphi}^{x, M}$. We have to find a curve $\gamma: I \rightarrow M$ such that $(\varphi \circ \gamma)^{\prime}(0)=p$ and $\gamma(0)=x$. Therefore, a reasonable ansatz is given by $\varphi \circ \gamma(t)=t p+\varphi(x)$. Indeed, the equivalence class $\xi$ of the curve

$$
\gamma(t):=\varphi^{-1}(t p+\varphi(x))
$$

fulfills $\vartheta_{\varphi}^{x, M}(\xi)=p$. So, $\vartheta_{\varphi}^{x, M}$ is surjective and in total bijective.
This function allows us to equip $T_{x} M$ with a vector space structure over the scalar field $\mathbb{R}$. Multiplication of a tangent vector $\xi \in T_{x} M$ with a scalar $s \in \mathbb{R}$ is defined by taking the image of $\xi$ in $\mathbb{R}^{n}$ under $\vartheta_{\varphi}^{x, M}$, multiplying the result with $s$ and bring it back to $T_{x} M$ with the inverse of the bijection $\vartheta_{\varphi}^{x, M}$, i.e.

$$
\begin{equation*}
s \xi:=\left(\vartheta_{\varphi}^{x, M}\right)^{-1}\left(s \vartheta_{\varphi}^{x, M}(\xi)\right) . \tag{2.5}
\end{equation*}
$$

The addition for $\xi, \eta \in T_{x} M$ is defined similarly over the addition in $\mathbb{R}^{n}$ via

$$
\xi+\eta:=\left(\vartheta_{\varphi}^{x, M}\right)^{-1}\left(\vartheta_{\varphi}^{x, M}(\xi)+\vartheta_{\varphi}^{x, M}(\eta)\right) .
$$

Remark 2.1.24. These two operations are well-defined, i.e. independent of the chart $(U, \varphi)$. To see this, let's take a different chart $(V, \psi)$ of $M$. According to [37, Section 2.6.1] the function $\vartheta_{\psi}^{x, M} \circ\left(\vartheta_{\varphi}^{x, M}\right)^{-1}$ is linear (see also [37, Section 1.3]). So we can do the following equivalence transformations omitting the indices $x$ and $M$

$$
\begin{aligned}
\vartheta_{\varphi}^{-1}\left(\vartheta_{\varphi}(\xi)+\vartheta_{\varphi}(\eta)\right) & =\vartheta_{\psi}^{-1}\left(\vartheta_{\psi}(\xi)+\vartheta_{\psi}(\eta)\right) \\
\vartheta_{\psi} \circ \vartheta_{\varphi}^{-1}\left(\vartheta_{\varphi}(\xi)+\vartheta_{\varphi}(\eta)\right) & =\vartheta_{\psi}(\xi)+\vartheta_{\psi}(\eta) \\
\vartheta_{\psi} \circ \vartheta_{\varphi}^{-1}\left(\vartheta_{\varphi}(\xi)\right)+\vartheta_{\psi} \circ \vartheta_{\varphi}^{-1}\left(\vartheta_{\varphi}(\eta)\right) & =\vartheta_{\psi}(\xi)+\vartheta_{\psi}(\eta) \\
\vartheta_{\psi}(\xi)+\vartheta_{\psi}(\eta) & =\vartheta_{\psi}(\xi)+\vartheta_{\psi}(\eta)
\end{aligned}
$$

that prove that this addition is independent of the chart. Analogously, scalar multiplication as defined above is well-defined.
Furthermore, the vector space axioms transfer directly from $\mathbb{R}^{n}[37$, Section 2.6.1].

### 2.1.3. The Special Orthogonal Group $\mathrm{SO}_{n}$

After the preparations of Section 2.1.2, we can investigate the geometric structure of the special orthogonal group $\mathrm{SO}_{n}$. We will see that $\mathrm{O}_{n}$ is a submanifold of $\mathbb{R}^{n \times n}$ and conclude that $\mathrm{SO}_{n}$ is a smooth manifold. Furthermore, we investigate the tangent space of $\mathrm{SO}_{n}$.

We start with a lemma about the representation of elements of $\mathrm{SO}_{n}$ with orthogonal matrices. The lemma is based on [21, Theorem 12.10] and parts of [21, Theorem 18.1], and will be used later.

Lemma 2.1.25. Any matrix $R \in \mathrm{SO}_{n}$ can be represented in the form $R=P B P^{T}$ with an orthogonal matrix $P \in \mathrm{O}_{n}$ and a block diagonal matrix

$$
\begin{equation*}
B=\operatorname{diag}\left(R_{1}\left(\alpha_{1}\right), R_{2}\left(\alpha_{2}\right), \ldots, R_{m}\left(\alpha_{m}\right), 1, \ldots, 1\right) \tag{2.6}
\end{equation*}
$$

where $R_{i}\left(\alpha_{i}\right) \in \mathrm{SO}_{2}$ for $i=1, \ldots, m$ denotes a rotation matrix

$$
R_{i}\left(\alpha_{i}\right)=\left(\begin{array}{rr}
\cos \alpha_{i} & -\sin \alpha_{i} \\
\sin \alpha_{i} & \cos \alpha_{i}
\end{array}\right) \quad \text { with } 0<\alpha_{i} \leq \pi
$$

Proof. Let $R \in \mathrm{SO}_{n}$ be a rotation matrix. The matrix $R$ is orthogonal, so [21, Theorem 12.10] states that there exist an orthogonal matrix $P \in \mathrm{O}_{n}$ and a block diagonal matrix $B=\operatorname{diag}\left(B_{1}, B_{2}, \ldots, B_{r}\right)$ where the blocks $B_{j}$ are either $1,-1$, or of the form

$$
B_{j}=\left(\begin{array}{rr}
\cos \theta_{j} & -\sin \theta_{j}  \tag{2.7}\\
\sin \theta_{j} & \cos \theta_{j}
\end{array}\right) \quad \text { with } 0<\theta_{j}<\pi
$$

with $R=P B P^{T}$ for $j=1, \ldots, r$. Since $R \in \mathrm{SO}_{n}$, the determinant

$$
\operatorname{det}(R)=\operatorname{det}\left(P B P^{T}\right)=\operatorname{det}(P) \operatorname{det}(B) \operatorname{det}\left(P^{T}\right)=\operatorname{det}(P)^{2} \operatorname{det}(B)^{P \in \mathrm{O}_{n}}=\operatorname{det}(B)
$$

has to be +1 . The determinant of the block diagonal matrix $B$ is given by the product $\operatorname{det}\left(B_{1}\right) \operatorname{det}\left(B_{2}\right) \cdots \operatorname{det}\left(B_{r}\right)$. Blocks of the form (2.7) are in $\mathrm{SO}_{2}$ and therefore have determinant 1. Thus, there is an even number of blocks of $B$ that are -1 . Hence we can assume, that $B$ has the form $\operatorname{diag}\left(\tilde{B}_{1}, \tilde{B}_{2}, \ldots, \tilde{B}_{\tilde{r}}\right)$ with blocks $\tilde{B}_{j}$ that are either 1 or of the form

$$
\tilde{B}_{j}=\left(\begin{array}{rr}
\cos \tilde{\theta}_{j} & -\sin \tilde{\theta}_{j}  \tag{2.8}\\
\sin \tilde{\theta}_{j} & \cos \tilde{\theta}_{j}
\end{array}\right) \quad \text { with } 0<\tilde{\theta}_{j} \leq \pi,
$$

for $j=1, \ldots, \tilde{r}$, where two -1 entries create such a $(2 \times 2)$-block, since $\cos \pi=-1$ and $\sin \pi=0$. The blocks of $B$ can be reordered by switching rows and columns with some orthogonal matrix $S$ similar to

$$
\underbrace{\left(\begin{array}{lll}
0 & 1 & 0 \\
0 & 0 & 1 \\
1 & 0 & 0
\end{array}\right)}_{=S}\left(\begin{array}{lll}
1 & 0 & 0 \\
0 & a & b \\
0 & c & d
\end{array}\right) \underbrace{\left(\begin{array}{lll}
0 & 0 & 1 \\
1 & 0 & 0 \\
0 & 1 & 0
\end{array}\right)}_{=S^{T}}=\left(\begin{array}{lll}
a & b & 0 \\
c & d & 0 \\
0 & 0 & 1
\end{array}\right)
$$

at the correct positions. After that, $B$ has the form (2.6), and the lemma is proven.
In the following, we want to show that $\mathrm{O}_{n}$ and consequently $\mathrm{SO}_{n}$ are smooth manifolds. Therefore we give a version of a statement of differential geometry known as rank theorem, regular level set theorem, or as a theorem about implicitly defined manifolds that can be found in various books on manifolds or calculus like [59, Theorem 9.9], [40, Chapter 5], and [1, Proposition 3.3.3]. We refer to these sources for a proof of that statement.
Lemma 2.1.26. Let $M$ and $N$ be two smooth manifolds of dimensions $m$ and $n$ with $m \geq n$. Furthermore, let $F: M \rightarrow N$ be a smooth function and $y \in N$ be a point on the manifold $N$. The point $y$ is called a regular value of $F$ if $F$ has full rank at every point $x \in F^{-1}(y)$, i.e. if $D F(x)[\cdot]$ is surjective at every $x \in F^{-1}(y)$. If $y \in N$ is a regular value of $F$, the pre-image $F^{-1}(y)$ is a submanifold of $M$ of dimension $m-n$.

Now we consider the orthogonal group $\mathrm{O}_{n} \subseteq \mathbb{R}^{n \times n}$. Example 2.1.21 examined that $\mathbb{R}^{n \times n}$ is an $n^{2}$-dimensional smooth manifold. Analogously, one can prove that any finitedimensional vector space $V$ over $\mathbb{R}$ can be equipped with an atlas such that $V$ is a smooth manifold. In particular, the set $\operatorname{Sym}_{n} \subseteq \mathbb{R}^{n \times n}$ of symmetric $(n \times n)$-matrices, i.e. $A=A^{T}$, is a smooth manifold. The following considerations are based on [1, Section 3.3.2].
Theorem 2.1.27. The orthogonal group $\mathrm{O}_{n}$ is a smooth manifold of dimension $\frac{n(n-1)}{2}$.
Proof. We will show that $\mathrm{O}_{n}$ is an embedded submanifold of $\mathbb{R}^{n \times n}$. Consider the function

$$
F: \mathbb{R}^{n \times n} \rightarrow \operatorname{Sym}_{n}: A \mapsto A^{T} A-I_{n} .
$$

Since $\left(A^{T} A\right)^{T}=A^{T} A$ for all $A \in \mathbb{R}^{n \times n}$, the function $F$ is well-defined. According to Definition 2.1.1, it holds that $\mathrm{O}_{n}=F^{-1}\left(\left\{0_{n}\right\}\right)$, where $0_{n}$ denotes the zero matrix in $\mathbb{R}^{n \times n}$.

Consider the differential $D F(A)[B]$ of $F$ at $A$ in the direction of $B$. With the Leibniz rule for differentiation, it holds (see [1, Appendix A.5] for details about matrix differentiation)

$$
D F(A)[B]=A^{T} B+B^{T} A .
$$

The mapping $D F(A)[\cdot]$ is surjective for every $A \in \mathrm{O}_{n}$ if for every $C \in \operatorname{Sym}_{n}$ there exists a matrix $B \in \mathbb{R}^{n \times n}$ with $D F(A)[B]=C$. For $A \in F^{-1}\left(\left\{0_{n}\right\}\right)=\mathrm{O}_{n}$ and $C \in \operatorname{Sym}_{n}$, let $B=\frac{1}{2} A C$, resulting in

$$
D F(A)\left[\frac{1}{2} A C\right]=A^{T} \frac{1}{2} A C+\left(\frac{1}{2} A C\right)^{T} A=\frac{1}{2}\left(A^{T} A C+C^{T} A^{T} A\right)=C
$$

with $A^{T} A=I_{n}$, since $A \in \mathrm{O}_{n}$, and $C=C^{T}$, since $C \in \operatorname{Sym}_{n}$. Thus, $0_{n}$ is a regular value of $F$ and therefore $\mathrm{O}_{n}$ a submanifold of $\mathbb{R}^{n \times n}$ with Lemma 2.1.26. The vector space $\mathrm{Sym}_{n}$ is $\frac{n(n+1)}{2}$-dimensional since for a symmetric matrix $S$, every element of the diagonal of $S$ and every element above this diagonal can be chosen independently. Hence, the dimension of $\mathrm{O}_{n}$ is given by $n^{2}-\frac{n(n+1)}{2}=\frac{n(n-1)}{2}$.

It follows that $\mathrm{SO}_{n}$ is a manifold as well if we can show that it is a connected component of $\mathrm{O}_{n}$. A connected component of a topological space $X$ is a subset $C \subseteq X$ that is connected and there exists no larger connected set $D \supsetneq C$ in $X$ [47, Chapter $3 \S 25$ ].
Theorem 2.1.28. The special orthogonal group $\mathrm{SO}_{n}$ is a smooth manifold of dimension $\frac{n(n-1)}{2}$.
Proof. We show that $\mathrm{SO}_{n}$ is a connected component of $\mathrm{O}_{n}$ as in [61]. Let $A$ be any matrix in $\mathrm{SO}_{n}$. The goal is to find an arc in $\mathrm{SO}_{n}$ that starts from $I_{n}$ and ends in $A$. With Lemma 2.1.25, the matrix $A$ can be represented in the form $A=P B P^{T}$ with an orthogonal matrix $P \in \mathrm{O}_{n}$ and a block diagonal matrix

$$
B=\operatorname{diag}\left(R_{1}\left(\alpha_{1}\right), R_{2}\left(\alpha_{2}\right), \ldots, R_{m}\left(\alpha_{m}\right), 1, \ldots, 1\right),
$$

for rotations $\mathbb{R}_{i}\left(\alpha_{i}\right) \in \mathrm{SO}_{2}$ by the angle $\alpha_{i} \in(0, \pi]$. For a $\lambda \in[0,1]$, let $B(\lambda)$ define the block diagonal matrix

$$
B(\lambda)=\operatorname{diag}\left(R_{1}\left(\lambda \alpha_{1}\right), R_{2}\left(\lambda \alpha_{2}\right), \ldots, R_{m}\left(\lambda \alpha_{m}\right), 1, \ldots, 1\right)
$$

and $A(\lambda)=P B(\lambda) P^{T}$. Clearly, $A(\lambda) \in \mathrm{SO}_{n}$ for all $\lambda \in[0,1]$. Furthermore, this arc starts at $A(0)=P I_{n} P^{T}=I_{n}$ and ends at $B(1)=A$. So the special orthogonal group $\mathrm{SO}_{n}$ is connected. Furthermore, since the function det: $\mathrm{O}_{n} \rightarrow\{-1,1\}$ is continuous, there cannot be a continuous path in $\mathrm{O}_{n}$ from a matrix $C \in \mathrm{O}_{n}$ with $\operatorname{det} C=1$ to a matrix $D \in \mathrm{O}_{n}$ with det $D=-1$. Hence, $\mathrm{SO}_{n}$ is a connected component of $\mathrm{O}_{n}$ and therefore open.

Thus, $\mathrm{SO}_{n}$ is an $\frac{n(n-1)}{2}$-dimensional topological manifold, since every point $x \in \mathrm{SO}_{n}$ has an open neighborhood $U_{x} \subseteq \mathrm{O}_{n}$ in $\mathrm{O}_{n}$ and therefore an open neighborhood $U_{x} \cap \mathrm{SO}_{n}$ in $\mathrm{SO}_{n}$ that is homeomorphic to some subset of $\mathbb{R}^{\frac{n(n-1)}{2}}$. In the same way, charts and atlases transfer from the smooth manifold $\mathrm{O}_{n}$ to $\mathrm{SO}_{n}$. Consequently, $\mathrm{SO}_{n}$ is a smooth manifold of dimension $\frac{n(n-1)}{2}$.

We want to investigate the tangent space to the manifold $\mathrm{SO}_{n}$ as in [21, Section 14.7]. At the identity $I_{n}$, the tangent vectors are given by the curves $\gamma: I \rightarrow \mathrm{SO}_{n}$ with $\gamma(0)=I_{n}$, w.l.o.g. let $I=(-1,1)$. Since $\gamma(t)$ denotes a matrix in $\mathrm{SO}_{n}$, we know $\gamma(t) \gamma(t)^{T}=I_{n}$. We can differentiate $\gamma$ with respect to $t$ resulting in

$$
\gamma^{\prime}(t) \gamma(t)^{T}+\gamma(t) \gamma^{\prime}(t)^{T}=0_{n}
$$

with the product rule. Since $\gamma(0)=I_{n}$, this reduces to

$$
\gamma^{\prime}(0)+\gamma^{\prime}(0)^{T}=0_{n} .
$$

Thus, $\gamma^{\prime}(0)$ is a skew-symmetric matrix. With $\mathrm{Skew}_{n}$ we denote the set of all skewsymmetric matrices in $\mathbb{R}^{n \times n}$. This is a $\frac{n(n-1)}{2}$-dimensional vector space since every element above the diagonal of a matrix can be chosen arbitrarily. We have seen above, that $T_{I_{n}} \mathrm{SO}_{n}$ is a $\frac{n(n-1)}{2}$-dimensional vector space over $\mathbb{R}$ as well. Hence, the spaces $\mathrm{Skew}_{n}$ and $T_{I_{n}} \mathrm{SO}_{n}$ are equal (more precisely, they can be identified).

At some arbitrary point $B \in \mathrm{SO}_{n}$, let's consider some curve $\gamma_{B}:(-1,1) \rightarrow \mathrm{SO}_{n}$ with $\gamma_{B}(0)=B$. Then the curve $\tilde{\gamma}_{B}(t):=B^{T} \gamma_{B}(t)$ passes through $I_{n}$ at 0 . So as seen above, we can write

$$
\tilde{\gamma}_{B}^{\prime}(0)=B^{T} \gamma_{B}^{\prime}(0) \in T_{I_{n}} \mathrm{SO}_{n}=\operatorname{Skew}_{n}
$$

and therefore

$$
\begin{equation*}
T_{B} \mathrm{SO}_{n}=\left\{B S \mid S \in \mathrm{Skew}_{n}\right\} \tag{2.9}
\end{equation*}
$$

We have seen above that for two manifolds $M_{1}$ and $M_{2}$ the product $M_{1} \times M_{2}$ is a manifold. This directly implies the following theorem.

Theorem 2.1.29. The special Euclidean group $\mathrm{SE}_{n}$ is a smooth manifold of dimension $\frac{n(n+1)}{2}$.

Proof. With (2.3) the special Euclidean group can be identified with $\mathrm{SO}_{n} \times \mathbb{R}^{n}$. According to Theorem 2.1.28, $\mathrm{SE}_{n}$ is the product of two manifolds of dimensions $\frac{n(n-1)}{2}$ and $n$. Thus, $\mathrm{SE}_{n}$ is a smooth manifold of dimension $\frac{n(n+1)}{2}$.

In the analysis of the pose estimation algorithm, we want to measure the quality of the estimation as the distance of the estimated poses to their ground truth. We will simplify the regarded objects to points in $\mathbb{R}^{3}$ with an orientation in $\mathrm{SO}_{3}$. Thus, the representation of the poses as elements of $\mathrm{SE}_{3}$ suffices since a point with an orientation has no symmetry. We can define a simple metric on $\mathrm{SO}_{n}$ and $\mathrm{SE}_{n}$ according to [34, Section 3.5].

Lemma 2.1.30. The function

$$
d_{\mathrm{SO}_{n}}: \mathrm{SO}_{n} \times \mathrm{SO}_{n} \rightarrow \mathbb{R}_{0}^{+}:\left(R_{1}, R_{2}\right) \mapsto\left\|I_{3}-R_{1} R_{2}^{T}\right\|_{F}
$$

defines a metric on $\mathrm{SO}_{n}$, where $\|A\|_{F}$ denotes the Frobenius norm for quadratic matrices.
Proof. The Frobenius norm $\|A\|_{F}$ of a matrix $A \in \mathbb{R}^{n \times n}$ is defined as

$$
\|A\|_{F}:=\sqrt{\sum_{i, j=1}^{n}\left|a_{i j}\right|^{2}}=\sqrt{\operatorname{tr}\left(A A^{T}\right)}
$$

and according to [58, Section I.3] a norm. Therefore, $d_{\mathrm{SO}_{n}}\left(R_{1}, R_{2}\right)=\left\|I_{n}-R_{1} R_{2}^{T}\right\|_{F}$ is positive or zero. In particular, $\left\|I_{n}-R_{1} R_{2}^{T}\right\|_{F}=0$ if and only if $I_{n}-R_{1} R_{2}^{T}=0_{n}$. This is equivalent to $I_{n}=R_{1} R_{2}^{T}$, and, since $R_{2}$ is in $\mathrm{O}_{n}$, this is equivalent to $R_{1}=R_{2}$. We conclude that $d_{\mathrm{SO}_{n}}\left(R_{1}, R_{2}\right)=0$ if and only if $R_{1}=R_{2}$.

The computation

$$
\left\|I_{n}-R_{1} R_{2}^{T}\right\|_{F}=\left\|\left(I_{n}-R_{1} R_{2}^{T}\right)^{T}\right\|_{F}=\left\|I_{n}-R_{2} R_{1}^{T}\right\|_{F}
$$

implies that $d_{\mathrm{SO}_{n}}\left(R_{1}, R_{2}\right)=d_{\mathrm{SO}_{n}}\left(R_{2}, R_{1}\right)$.
To proof the triangle inequality $d_{\mathrm{SO}_{n}}\left(R_{1}, R_{2}\right) \leq d_{\mathrm{SO}_{n}}\left(R_{1}, R_{3}\right)+d_{\mathrm{SO}_{n}}\left(R_{3}, R_{2}\right)$ for any matrix $R_{3} \in \mathrm{SO}_{n}$, we start by proving $\left\|I_{n}-R S^{T}\right\|_{F}=\|S-R\|_{F}$ for all $R, S \in \mathrm{SO}_{n}$. With $\|A\|_{F}^{2}=\operatorname{tr}\left(A A^{T}\right)$ and $R R^{T}=S S^{T}=I_{n}$, we compute

$$
\begin{aligned}
\|S-R\|_{F}^{2} & =\operatorname{tr}\left((S-R)(S-R)^{T}\right) \\
& =\operatorname{tr}\left(S S^{T}-R S^{T}-S R^{T}+R R^{T}\right) \\
& =\operatorname{tr}\left(I_{n} I_{n}^{T}-\left(R S^{T}\right) I_{n}^{T}-I_{n}\left(R S^{T}\right)^{T}+I_{n} I_{n}^{T}\right) \\
& =\operatorname{tr}\left(\left(I_{n}-R S^{T}\right)\left(I_{n}-R S^{T}\right)^{T}\right) \\
& =\left\|I_{n}-R S^{T}\right\|_{F}^{2} .
\end{aligned}
$$

Thus, we conclude

$$
\begin{aligned}
d_{\mathrm{SO}_{n}}\left(R_{1}, R_{2}\right) & =\left\|I_{n}-R_{1} R_{2}^{T}\right\|_{F} \\
& =\left\|R_{2}-R_{1}\right\|_{F} \\
& =\left\|R_{2}-R_{3}+R_{3}-R_{1}\right\|_{F} \\
& \leq\left\|R_{2}-R_{3}\right\|_{F}+\left\|R_{3}-R_{1}\right\|_{F} \\
& =\left\|I_{n}-R_{3} R_{2}^{T}\right\|_{F}+\left\|I_{n}-R_{1} R_{3}^{T}\right\|_{F} \\
& =d_{\mathrm{SO}_{n}}\left(R_{1}, R_{3}\right)+d_{\mathrm{SO}_{n}}\left(R_{3}, R_{2}\right) .
\end{aligned}
$$

So the triangle inequality holds for $d_{\mathrm{SO}_{n}}$ and therefore $d_{\mathrm{SO}_{n}}$ is a metric.
Theorem 2.1.31. The function

$$
d_{\mathrm{SE}_{n}}: \mathrm{SE}_{n} \times \mathrm{SE}_{n} \rightarrow \mathbb{R}_{0}^{+}:\left(T_{1}, T_{2}\right)=\left(\left(R_{1}, t_{1}\right),\left(R_{2}, t_{2}\right)\right) \mapsto d_{\mathrm{SO}_{n}}\left(R_{1}, R_{2}\right)+d_{2}\left(t_{1}, t_{2}\right),
$$

where $d_{2}$ is the Euclidean metric on $\mathbb{R}^{n}$, is a metric on $\mathrm{SE}_{n}$.
Proof. With Lemma 2.1.30, $d_{\mathrm{SO}_{n}}$ is a metric on $\mathrm{SO}_{n}$. Therefore, positivity, symmetry, and the triangle inequality transfer from the metrics $d_{\mathrm{SO}_{n}}$ and $d_{2}$ directly to $d_{\mathrm{SE}_{n}}$.

In the following, we will omit the indices $\mathrm{SO}_{n}$ and $\mathrm{SE}_{n}$ if it is clear which metric is used.

### 2.1.4. The Exponential Map for Quadratic Matrices

As discussed at the beginning of Section 2.1.2, it is not possible to make "small steps" towards an optimal solution on the $\mathrm{SO}_{3}$ manifold by adding an arbitrary small matrix $V \in \mathbb{R}^{3 \times 3}$ to a rotation matrix $R \in \mathrm{SO}_{3}$. But if $V$ has a certain structure, it can easily be brought back onto $\mathrm{SO}_{3}$ using the exponential map for quadratic matrices [13, Section 6.1].
This exponential map is defined analogously to the power series of the exponential function on $\mathbb{C}$. Here the definition as well as the proof of the well-definedness and basic calculation rules are based on [21, Section 18.1]

Definition 2.1.32. The exponential $e^{A}$ (sometimes denoted as $\exp A$ ) of a quadratic ma$\operatorname{trix} A \in \mathbb{C}^{n \times n}$ is defined as

$$
\begin{equation*}
e^{A}=\sum_{k \geq 0} \frac{A^{k}}{k!} \tag{2.10}
\end{equation*}
$$

with the matrix powers $A^{k}=A^{k-1} A$ and $A^{0}:=I$ for all $A$.
To prove that $e^{A}$ is well-defined, we first need a lemma about the powers of matrices.
Lemma 2.1.33. Let $A \in \mathbb{C}^{n \times n}$ be a real quadratic matrix and let $a_{\max }$ be the maximum value of the absolute values $\left|a_{i j}\right|$ of all entries of $A$. Then for a positive integer $k \in \mathbb{Z}^{+}$, the absolute values of all entries $a_{i j}^{(k)}$ of the matrix $A^{k}$ are bounded by $\left(n a_{\max }\right)^{k}$.
Proof. Let $a_{\max }:=\max _{1 \leq i, j \leq n}\left|a_{i j}\right|$ be defined as stated above. We prove this lemma by induction on the exponent $k$ of $A$. For $k=1$, this statement is trivially true since $a_{i j}^{(1)}=a_{i j}$.

Assuming $\left|a_{i j}^{(k)}\right| \leq\left(n a_{\max }\right)^{k}$ is true for $k \in \mathbb{Z}^{+}$, consider $A^{k+1}$. Using $A^{k+1}=A^{k} A$, we can conclude for an entry $a_{i j}^{(k+1)}$ of $A^{k+1}$

$$
\begin{aligned}
\left|a_{i j}^{(k+1)}\right| & =\left|\sum_{1 \leq m \leq n} a_{i m}^{(k)} a_{m j}\right| \\
& \stackrel{(1)}{\leq} \sum_{1 \leq m \leq n}\left|a_{i m}^{(k)}\right|\left|a_{m j}\right| \\
& \stackrel{(2)}{\leq} \sum_{1 \leq m \leq n}\left(n a_{\max }\right)^{k} a_{\max } \\
& =n^{k} \sum_{1 \leq m \leq n} a_{\max }^{k+1} \\
& =n^{k} n a_{\max }^{k+1}=\left(n a_{\max }\right)^{k+1}
\end{aligned}
$$

For the inequality in (1), we use the triangle inequality and for the inequality in (2), we use the induction hypothesis.

Thus, $\left|a_{i j}^{(k+1)}\right| \leq\left(n a_{\max }\right)^{k+1}$ holds for all $k \in \mathbb{Z}^{+}$.
Lemma 2.1.34. The exponential map $e^{A}$ of a matrix $A \in \mathbb{C}^{n \times n}$ as presented in Definition 2.1.32 is well-defined, i.e. the power series (2.10) converges absolutely for all $A \in \mathbb{C}^{n \times n}$. Proof. We say the matrix power series (2.10) converges absolutely if each entry of the matrix sequence

$$
\sum_{k=0}^{N}\left|\frac{A^{k}}{k!}\right|
$$

converges for $N \rightarrow \infty$. With the notation and result of Lemma 2.1.33 we can write for the series of one entry

$$
\sum_{k \geq 0} \frac{\left|a_{i j}^{(k)}\right|}{k!} \leq \sum_{k \geq 0} \frac{\left(n a_{\max }\right)^{k}}{k!}=e^{n a_{\max }}
$$

and thus the series (2.10) converges absolutely due to the direct comparison test.
The exponential map has some interesting properties [21, Section 18.1].
Lemma 2.1.35. Let $A \in \mathbb{R}^{n \times n}$ and $B, C \in \mathbb{C}^{n \times n}$ be quadratic matrices.
(i) If $B$ and $C$ commute, i.e. $B C=C B$, then

$$
e^{B} e^{C}=e^{B+C}
$$

(ii) Let $P$ be a regular matrix in $\mathrm{GL}_{n}(\mathbb{R})$ or in $\mathrm{GL}_{n}(\mathbb{C})$ and let $D$ be a matrix in $\mathbb{R}^{n \times n}$ or $\mathbb{C}^{n \times n}$. Then, the equation

$$
e^{P D P^{-1}}=P e^{D} P^{-1}
$$

holds.
(iii) For the determinant of $e^{A}$ holds

$$
\operatorname{det}\left(e^{A}\right)=e^{\operatorname{tr} A}
$$

where $\operatorname{tr} A$ denotes the trace $a_{11}+a_{22}+\cdots+a_{n n}$ of $A$.
(iv) The exponential of $A$ is regular, i.e. $e^{A} \in \mathrm{GL}_{n}$.
(v) The inverse of $e^{A}$ is given by

$$
\left(e^{A}\right)^{-1}=e^{-A} .
$$

Proof. (i) To prove this property, consider the power series representations of $e^{B}$ and $e^{C}$ and their Cauchy product. In the following computation, we use the binomial formula for the equality in (1) which only holds since $B C=C B$, so

$$
\begin{aligned}
e^{B} e^{C} & =\left(\sum_{i \geq 0} \frac{B^{i}}{i!}\right)\left(\sum_{j \geq 0} \frac{C^{j}}{j!}\right) \\
& =\sum_{k \geq 0} \sum_{l=0}^{k} \frac{B^{l}}{l!} \frac{C^{k-l}}{(k-l)!} \\
& =\sum_{k \geq 0} \frac{1}{k!} \sum_{l=0}^{k}\binom{k}{l} B^{l} C^{k-l} \\
& \stackrel{(1)}{=} \sum_{k \geq 0} \frac{1}{k!}(B+C)^{k}=e^{B+C} .
\end{aligned}
$$

This proves the statement.
(ii) Since $P P^{-1}=I$ and $\left(P D^{k-1} P^{-1}\right)\left(P D P^{-1}\right)=P D^{k} P^{-1}$, induction shows that

$$
\left(P D P^{-1}\right)^{k}=P D^{k} P^{-1}
$$

for every $k \geq 0$. Thus, we see

$$
e^{P D P^{-1}}=\sum_{k \geq 0} \frac{\left(P D P^{-1}\right)^{k}}{k!}=\sum_{k \geq 0} \frac{P D^{k} P^{-1}}{k!}=P e^{D} P^{-1} .
$$

So $e^{P D P^{-1}}=P e^{D} P^{-1}$ is shown.
(iii) The characteristic polynomial $\chi_{A}$ of the matrix $A$ decomposes into linear factors over $\mathbb{C}$ due to the fundamental theorem of algebra. So according to [32, Satz 8.7.10] the matrix $A$ interpreted as an element of $\mathbb{C}^{n \times n}$ is similar to a matrix $J$ in Jordan normal form, i.e. there exists a matrix $P \in \mathrm{GL}_{n}(\mathbb{C})$ with $A=P J P^{-1}$.
Part (ii) implies that

$$
e^{A}=e^{P J P^{-1}}=P e^{J} P^{-1} .
$$

The Jordan matrix $J$ is an upper triangular matrix with the (complex) eigenvalues $\lambda_{i}$ for $i=1, \ldots, n$ of $A$ in its diagonal (according to their algebraic multiplicity). Again, a simple induction shows that for any upper triangular matrix $T=\left(t_{i j}\right)$ the matrix $e^{T}$ is also an upper triangular matrix, with the diagonal entries $e^{t_{i i}}$. So the main diagonal of $e^{J}$ consists of the exponentials $e^{\lambda_{i}}$ of the eigenvalues $\lambda_{i}$ of $A$.
The determinant of a triangular matrix is the product of its diagonal entries. To sum up, the determinant of $e^{A}$ computes as

$$
\operatorname{det}\left(e^{A}\right)=\operatorname{det}\left(P e^{J} P^{-1}\right)=\operatorname{det}(P) \operatorname{det}(P)^{-1} \operatorname{det}\left(e^{J}\right)=e^{\lambda_{1}} e^{\lambda_{2}} \cdots e^{\lambda_{n}}=e^{\lambda_{1}+\lambda_{2}+\cdots+\lambda_{n}},
$$

where the multiplicativity of the determinant and $\operatorname{det}\left(P^{-1}\right)=\operatorname{det}(P)^{-1}$ was used.
The trace $\operatorname{tr} A$ of the matrix $A$ is defined as the sum $a_{11}+a_{22}+\cdots+a_{n n}$ of the diagonal entries of $A$. Furthermore, the $\operatorname{trace} \operatorname{tr} A$ is equal to the sum of the eigenvalues of $A$ [33, Section 1.2]. This is no contradiction to complex eigenvalues: Since $\chi_{A}$ is a polynomial with real coefficients, for every zero $u=a+i b$ of $\chi_{A}$ that lies in $\mathbb{C} \backslash \mathbb{R}$, the complex conjugate $\bar{u}=a-i b$ is a zero of $\chi_{A}$, too. The sum $u+\bar{u}=2 a$ is in $\mathbb{R}$, therefore the sum of all eigenvalues of $A$ is in $\mathbb{R}$.
In total, we have

$$
\operatorname{det}\left(e^{A}\right)=e^{\lambda_{1}+\lambda_{2}+\cdots+\lambda_{n}}=e^{\operatorname{tr} A},
$$

which completes the proof.
(iv) According to (iii) the determinant of $e^{A}$ is equal to $e^{\operatorname{tr} A}$ which is positive. Therefore, $e^{A}$ is regular for any $A \in \mathbb{R}^{n \times n}$.
(v) As we learned in (iv), the matrix $e^{A}$ is regular, which means that it has an inverse. The matrices $A$ and $-A$ commute, since $A(-A)=-A^{2}=(-A) A$. Thus, with (i) follows

$$
e^{A} e^{-A}=e^{A-A}=e^{0_{n}}=I_{n} .
$$

A similar computation delivers $e^{-A} e^{A}=I_{n}$ which shows that $e^{-A}$ is the inverse matrix of $e^{A}$.

A matrix $A \in \mathbb{R}^{n \times n}$ is skew-symmetric if $A=-A^{T}$. We can derive an explicit formula for $e^{A}$ if $A$ is a skew-symmetric matrix in $\mathbb{R}^{2 \times 2}$.
Theorem 2.1.36. Let $A \in \mathbb{R}^{2 \times 2}$ be a skew-symmetric matrix of the form

$$
A=\left(\begin{array}{rr}
0 & -\theta \\
\theta & 0
\end{array}\right) .
$$

Then the exponential of $A$ is given by

$$
e^{A}=\left(\begin{array}{rr}
\cos \theta & -\sin \theta \\
\sin \theta & \cos \theta
\end{array}\right) .
$$

Thus, $e^{A}$ is in $\mathrm{SO}_{2}$ [21, Section 18.1].
Proof. First, we take a look at the skew-symmetric matrix

$$
M:=\left(\begin{array}{rr}
0 & -1 \\
1 & 0
\end{array}\right) .
$$

By a simple induction with the induction start

$$
M^{2}=\left(\begin{array}{rr}
-1 & 0 \\
0 & -1
\end{array}\right), \quad M^{3}=\left(\begin{array}{rr}
0 & 1 \\
-1 & 0
\end{array}\right), \quad M^{4}=\left(\begin{array}{ll}
1 & 0 \\
0 & 1
\end{array}\right),
$$

and $A=\theta M$, we see for any integer $k \geq 0$

$$
\begin{aligned}
& A^{4 k+1}=\theta^{4 k+1}\left(\begin{array}{rr}
0 & -1 \\
1 & 0
\end{array}\right)=\theta^{4 k+1} M, \quad A^{4 k+2}=\theta^{4 k+2}\left(\begin{array}{rr}
-1 & 0 \\
0 & -1
\end{array}\right)=-\theta^{4 k+2} I_{2}, \\
& A^{4 k+3}=\theta^{4 k+3}\left(\begin{array}{rr}
0 & 1 \\
-1 & 0
\end{array}\right)=-\theta^{4 k+3} M, \quad A^{4 k+4}=\theta^{4 k+4}\left(\begin{array}{ll}
1 & 0 \\
0 & 1
\end{array}\right)=\theta^{4 k+4} I_{2} .
\end{aligned}
$$

In Lemma 2.1.34 we proved that the power series of $e^{A}$ converges absolutely. Thus, we can rearrange the terms. So, rearranging and using the series expansions of $\sin \theta$ and $\cos \theta$ gives

$$
\begin{aligned}
e^{A} & =\sum_{k \geq 0} \frac{A^{k}}{k!}=I_{3}+\frac{A}{1!}+\frac{A^{2}}{2!}+\frac{A^{3}}{3!}+\frac{A^{4}}{4!}+\frac{A^{5}}{5!}+\ldots \\
& =\left(I_{3}+\frac{1}{2!} A^{2}+\frac{1}{4!} A^{4}+\ldots\right)+\left(A+\frac{1}{3!} A^{3}+\frac{1}{5!} A^{5}+\ldots\right) \\
& =\left(\theta^{0} I_{2}+\frac{1}{2!}\left(-\theta^{2} I_{2}\right)+\frac{1}{4!} \theta^{4} I_{2}+\ldots\right)+\left(\theta M+\frac{1}{3!}\left(-\theta^{3} M\right)+\frac{1}{5!} \theta^{5} M+\ldots\right) \\
& =\left(\sum_{k \geq 0}(-1)^{k} \frac{\theta^{2 k}}{(2 k)!}\right) I_{2}+\left(\sum_{k \geq 0}(-1)^{k} \frac{\theta^{2 k+1}}{(2 k+1)!}\right) M \\
& =\cos \theta I_{2}+\sin \theta M=\left(\begin{array}{rr}
\cos \theta & -\sin \theta \\
\sin \theta & \cos \theta
\end{array}\right) .
\end{aligned}
$$

This computation finishes the proof.

This proves that the exponential $e^{A}$ of a skew-symmetric matrix $A \in \mathbb{R}^{2 \times 2}$ is a rotation matrix in $\mathrm{SO}_{2}$. This even holds for skew-symmetric matrices in $\mathbb{R}^{n \times n}$ for arbitrary $n \geq 2$. Gallier gives in [21, Theorem 18.1] the following statement.

Theorem 2.1.37. Let $\mathrm{Skew}_{n} \subseteq \mathbb{R}^{n \times n}$ be the set of all real skew-symmetric ( $n \times n$ )-matrices.
(i) For all $A \in \operatorname{Skew}_{n}$, the exponential $e^{A}$ is an element of $\mathrm{SO}_{n}$.
(ii) The exponential map

$$
\begin{equation*}
e^{\prime}: \mathrm{Skew}_{n} \rightarrow \mathrm{SO}_{n} \tag{2.11}
\end{equation*}
$$

with the domain $\mathrm{Skew}_{n}$ and the codomain $\mathrm{SO}_{n}$ is surjective.
Proof. (i) Let $A \in \mathrm{Skew}_{n}$ be a skew-symmetric matrix. First, we show that $A \in \mathrm{O}_{n}$. Since $A$ is skew-symmetric, we know $A^{T}=-A$. Considering the power series representation of $e^{A}$, we see that $\left(e^{A}\right)^{T}=e^{A^{T}}$. With Lemma 2.1.35(v) follows

$$
\left(e^{A}\right)^{T} e^{A}=e^{A^{T}} e^{A}=e^{-A} e^{A}=I_{n}
$$

and $e^{A}\left(e^{A}\right)^{T}=I_{n}$ analogously. Thus, the matrix $e^{A}$ is orthogonal, i.e.

$$
\left(e^{A}\right)^{-1}=\left(e^{A}\right)^{T} \text {. }
$$

Second, we compute the determinant $\operatorname{det}\left(e^{A}\right)$. The property $A^{T}=-A$ implies that the diagonal entries of $A$ are all equal to zero. Thus, $\operatorname{tr} A=0$. With Lemma 2.1.35(iii) follows

$$
\operatorname{det}\left(e^{A}\right)=e^{\operatorname{tr} A}=e^{0}=1
$$

To sum up, from $e^{A} \in \mathrm{O}_{n}$ and $\operatorname{det}\left(e^{A}\right)=1$ follows that $e^{A} \in \mathrm{SO}_{n}$.
(ii) Part (i) guarantees that the function (2.11) is well-defined. Let $R \in \mathrm{SO}_{n}$ be a rotation matrix. With Lemma 2.1.25, the matrix $R$ can be represented in the form $R=P B P^{T}$ with an orthogonal matrix $P \in \mathrm{O}_{n}$ and a block diagonal matrix

$$
B=\operatorname{diag}\left(B_{1}, B_{2}, \ldots, B_{r}, 1, \ldots, 1\right) .
$$

The blocks $B_{j} \in \mathrm{SO}_{2}$ are of the form

$$
B_{j}=\left(\begin{array}{rr}
\cos \theta_{j} & -\sin \theta_{j} \\
\sin \theta_{j} & \cos \theta_{j}
\end{array}\right) \quad \text { with } 0<\theta_{j} \leq \pi .
$$

We have to find a skew-symmetric matrix $A$ with $e^{A}=R$. Let's define a block diagonal matrix $C=\operatorname{diag}\left(C_{1}, C_{2}, \ldots, C_{r}, 0, \ldots, 0\right)$ as follows. For a block $B_{j}$ let $C_{j}$ be the $(2 \times 2)$-matrix

$$
C_{j}:=\left(\begin{array}{rr}
0 & -\theta_{j}  \tag{2.12}\\
\theta_{j} & 0
\end{array}\right) .
$$

The exponential $e^{C}$ of the block diagonal matrix $C$ can be computed by taking the exponential of the blocks separately as

$$
e^{C}=\operatorname{diag}\left(e^{C_{1}}, e^{C_{2}}, \ldots, e^{C_{r}}, e^{0}, \ldots, e^{0}\right)
$$

The exponential $e^{C_{j}}$ for $j=1, \ldots, r$ equals $B_{j}$ due to Theorem 2.1.36. Since $e^{0}=1$, we have $e^{C}=B$.
Now, let $A$ be the matrix $P C P^{T}$ with the orthogonal matrix $P$ from above. The matrix $C$ consists of skew-symmetric blocks on its diagonal. So $C$ is skew-symmetric, implying $C+C^{T}=0_{n}$. We can compute

$$
A+A^{T}=P C P^{T}+\left(P C P^{T}\right)^{T}=P C P^{T}+P C^{T} P^{T}=P\left(C+C^{T}\right) P^{T}=0_{n}
$$

which shows that $A$ is skew-symmetric. The exponential of $A$ can be rewritten as

$$
e^{A}=e^{P C P^{T} 2.1 .35(i i)}=P e^{C} P^{T}=P B P^{T}=R
$$

with the considerations above and Lemma 2.1.35(ii).
So, for an arbitrary rotation matrix $R \in \mathrm{SO}_{n}$, we found a skew-symmetric matrix $A \in \operatorname{Skew}_{n}$ such that $e^{A}=R$. Thus, the exponential map is surjective with domain Skew $_{n}$ and codomain $\mathrm{SO}_{n}$.

For $n=3$, Rodrigues' formula gives an explicit representation of $e^{A}$ for a skew-symmetric matrix $A$, allowing an efficient computation of $e^{A}$.

Theorem 2.1.38 (Rodrigues' Formula). Let $A \in \mathbb{R}^{3 \times 3}$ be a skew-symmetric matrix of the form

$$
A=\left(\begin{array}{rrr}
0 & -c & b \\
c & 0 & -a \\
-b & a & 0
\end{array}\right)
$$

and $\theta:=\sqrt{a^{2}+b^{2}+c^{2}}$. Then, for $A \neq 0_{3}$, Rodrigues' formula

$$
e^{A}=I_{3}+\frac{\sin \theta}{\theta} A+\frac{1-\cos \theta}{\theta^{2}} A^{2}
$$

holds [21, Lemma 18.6].
Proof. For a matrix $A$ as given above, consider the matrix

$$
\tilde{A}:=\left(\begin{array}{lll}
a^{2} & a b & a c \\
a b & b^{2} & b c \\
a c & b c & c^{2}
\end{array}\right) .
$$

The computation

$$
A \tilde{A}=\left(\begin{array}{rrr}
0 & -c & b \\
c & 0 & -a \\
-b & a & 0
\end{array}\right)\left(\begin{array}{rrr}
a^{2} & a b & a c \\
a b & b^{2} & b c \\
a c & b c & c^{2}
\end{array}\right)=\left(\begin{array}{rrr}
-c a b+b a c & -c b^{2}+b^{2} c & -c b c+b c^{2} \\
c a^{2}-a^{2} c & c a b-a b c & c a c-a c^{2} \\
-b a^{2}+a^{2} b & -b a b+a b^{2} & -b a c+a b c
\end{array}\right)=0_{3}
$$

and an analogous computation for $\tilde{A} A$ show that $A \tilde{A}=\tilde{A} A=0_{3}$. Furthermore, we can represent $A^{2}$ by $\tilde{A}$ and $\theta=\sqrt{a^{2}+b^{2}+c^{2}}$ via

$$
\begin{aligned}
A^{2} & =\left(\begin{array}{ccc}
-c^{2}-b^{2} & b a & c a \\
a b & -c^{2}-a^{2} & c b \\
a c & b c & -b^{2}-a^{2}
\end{array}\right) \\
& =\left(\begin{array}{ccc}
-c^{2}-b^{2}-a^{2}+a^{2} & a b & a c \\
a b & -c^{2}-a^{2}-b^{2}+b^{2} & b c \\
a c & b c & -b^{2}-a^{2}-c^{2}+c^{2}
\end{array}\right)=\tilde{A}-\theta^{2} I_{3} .
\end{aligned}
$$

Multiplying this equation by $A$ gives

$$
A^{3}=A\left(\tilde{A}-\theta^{2} I_{3}\right)=-\theta^{2} A
$$

since $A \tilde{A}=0_{3}$. It follows $A^{4}=-\theta^{2} A^{2}$. From these considerations, we can deduce for any positive integer $k$ by induction

$$
\begin{array}{ll}
A^{4 k+1}=\theta^{4 k} A & \mid \cdot A \\
A^{4 k+2}=\theta^{4 k} A^{2} & \mid \cdot A \\
A^{4 k+3}=\theta^{4 k} A^{3}=\theta^{4 k}\left(-\theta^{2} A\right)=-\theta^{4 k+2} A & \mid \cdot A \\
A^{4 k+4}=-\theta^{4 k+2} A^{2} &
\end{array}
$$

by assuming $A^{4 k}=-\theta^{4 k-2} A^{2}$ and using the identity shown above for $A^{3}$.
As in the $\mathbb{R}^{2 \times 2}$ case, we use the absolute convergence of $e^{A}$ to rearrange the terms of its power series and the power series expansions of $\sin \theta$ and $\cos \theta$. So we conclude

$$
\begin{aligned}
e^{A} & =\sum_{k \geq 0} \frac{A^{k}}{k!}=I_{3}+\frac{A}{1!}+\frac{A^{2}}{2!}+\frac{A^{3}}{3!}+\frac{A^{4}}{4!}+\frac{A^{5}}{5!}+\frac{A^{6}}{6!}+\ldots \\
& =I_{3}+\left(A+\frac{1}{3!} A^{3}+\frac{1}{5!} A^{5}+\ldots\right)+\left(\frac{1}{2!} A^{2}+\frac{1}{4!} A^{4}+\frac{1}{6!} A^{6}+\ldots\right) \\
& =I_{3}+\left(\theta^{0} A+\frac{1}{3!}\left(-\theta^{2} A\right)+\frac{1}{5!} \theta^{4} A+\ldots\right)+\left(\frac{1}{2!} \theta^{0} A^{2}+\frac{1}{4!}\left(-\theta^{2} A^{2}\right)+\frac{1}{6!} \theta^{4} A^{2}+\ldots\right) \\
& =I_{3}+\frac{1}{\theta}\left(\theta-\frac{1}{3!} \theta^{3}+\frac{1}{5!} \theta^{5}-+\ldots\right) A+\frac{1}{\theta^{2}}\left(1-1+\frac{1}{2!} \theta^{2}-\frac{1}{4!} \theta^{4}+\frac{1}{6!} \theta^{6}-+\ldots\right) A^{2} \\
& =I_{3}+\frac{1}{\theta}\left(\sum_{k \geq 0}(-1)^{k} \frac{\theta^{2 k+1}}{(2 k+1)!}\right) A+\frac{1}{\theta^{2}}\left(1-\sum_{k \geq 0}(-1)^{k} \frac{\theta^{2 k}}{(2 k)!}\right) A^{2} \\
& =I_{3}+\frac{\sin \theta}{\theta} A+\frac{1-\cos \theta}{\theta^{2}} A^{2} .
\end{aligned}
$$

This computation finishes the proof.

### 2.2. Factor Graphs

The idea of a complex object being composed of a few simpler objects directly translates to a graph-theoretic representation of this object. Each considered part of the composite object
as well as each (geometrical) relation between these parts is represented by a vertex in the object's graph model. The edges of the graph connect a vertex representing a part with all vertices representing the relations, that part is involved in. Since no two part-vertices and no two relation-vertices are connected by an edge, we receive a bipartite graph.

Example 2.2.1. Let's consider a simplified model of a house and its corresponding factor graph, constructed as described above. Let the house consist of four walls denoted as $w_{i}$ for $i=1,2,3,4$, a roof $r$, and a door $d$. A graph displaying this house could look like in Figure 2.1.


Figure 2.1.: Factor graph for a house viewed as a composite object
The round nodes represent the poses of the different parts of the house. The little black squares connecting two nodes $v_{1}$ and $v_{2}$ indicate the transformation from the pose of $v_{1}$ to the pose of $v_{2}$. These relative connections sufficiently describe the house as a composite object. If we want to describe the house in some greater context, i.e. in some world coordinate frame, we need some absolute conditions on a variable as well. In the pose estimation problem of Section 3.2, this is solved with an additional node, connected to only one node of a part of the composite object, called a prior factor.

This concept of representing a composite object with a graph is a key feature in our solution method of the pose estimation problem of Section 3.2, where the pose estimation scenario will be modeled with factor graphs as they are described in [36].

A factor graph consists of two disjoint sets of vertices, one referred to as variables and the other one referred to as factors. This leads to the formal definition of factor graphs.

Definition 2.2.2. Let $g: D \rightarrow R$ be a function with domain $D=A_{1} \times A_{2} \times \cdots \times A_{n}$ for some sets $A_{i}$ with $i=1, \ldots, n$ and any semiring $R$ as codomain. Suppose that there exist a finite index set $J$ and functions $f_{j}: D_{j} \rightarrow R$ for $j \in J$ with $D_{j}=A_{j_{1}} \times \cdots \times A_{j_{k}}$ and $j_{1}, \ldots, j_{k} \in\{1, \ldots, n\}$ pairwise different, such that the function $g$ factorizes as

$$
\begin{equation*}
g\left(x_{1}, \ldots, x_{n}\right)=\prod_{j \in J} f_{j}\left(X_{j}\right) \tag{2.13}
\end{equation*}
$$

with $X_{j}=\left(x_{j_{1}}, \ldots, x_{j_{k}}\right)$ where $f_{j}\left(X_{j}\right)$ stands for $f_{j}\left(x_{j_{1}}, \ldots, x_{j_{k}}\right)$. A factor graph for this factorization of $g$ is a graph $\mathcal{F}=(V, E)$ with the set of vertices $V=V_{v} \dot{\cup} V_{f}$ and the set of edges $E$ that is based on the factorization (2.13). The function $g$ is called the global function of the factor graph $\mathcal{F}$.

The nodes in $V_{v}$ are called variable nodes and represent the variables $x_{1}, \ldots, x_{n}$ of $g\left(x_{1}, \ldots, x_{n}\right)$, the nodes in $V_{f}$ are called factor nodes and represent the factors $f_{j}$ for $j \in J$ of the factorization of $g$. The set $E$ contains no edge between two variable nodes respectively two factor nodes. The factor node for a factor $f_{j}$ is connected to the variable node $x_{i}$ by an edge $\left\{f_{j}, x_{i}\right\} \in E$ if and only if $x_{i}$ is an argument of $f_{j}$.

So, every node in a factor graph is equipped with either a variable or a function. With the set of variable nodes (equipped with variables) $V_{v}$ and the set of factor nodes (equipped with functions) $V_{f}$, we also write $\mathcal{F}=\left(V_{v}, V_{f}, E\right)$ for the factor graph $\mathcal{F}$.


Figure 2.2.: Factor graph for the product $f_{a}\left(x_{1}\right) f_{b}\left(x_{2}, x_{4}\right) f_{c}\left(x_{1}, x_{2}, x_{5}\right) f_{d}\left(x_{3}, x_{4}, x_{5}\right)$.
Example 2.2.3. Let $g: \mathbb{Z}_{3}^{5} \rightarrow \mathbb{Z}_{3}$ be a function with the factorization

$$
g\left(x_{1}, x_{2}, x_{3}, x_{4}, x_{5}\right)=\underbrace{x_{1}^{2}}_{=: f_{a}\left(x_{1}\right)} \underbrace{\left(2 x_{2}+x_{4}+1\right)}_{=: f_{b}\left(x_{2}, x_{4}\right)} \underbrace{\left(x_{1}+x_{2}^{4}+x_{5}^{3}+2\right)}_{=: f_{c}\left(x_{1}, x_{2}, x_{5}\right)} \underbrace{\left(x_{3}+x_{4}+x_{5}\right)}_{=: f_{d}\left(x_{3}, x_{4}, x_{5}\right)} .
$$

With the index set $J=\{a, b, c, d\}$ and the respective factors, we can draw the factor graph in Figure 2.2. If we change the split of the function $g$ into factors to

$$
g\left(x_{1}, x_{2}, x_{3}, x_{4}, x_{5}\right)=\underbrace{x_{1}^{2}\left(2 x_{2}+x_{4}+1\right)}_{=: f_{e}\left(x_{1}, x_{2}, x_{4}\right)} \underbrace{\left(x_{1}+x_{2}^{4}+x_{5}^{3}+2\right)}_{=: f_{c}\left(x_{1}, x_{2}, x_{5}\right)} \underbrace{\left(x_{3}+x_{4}+x_{5}\right)}_{=: f_{d}\left(x_{3}, x_{4}, x_{5}\right)},
$$

we receive a different factor graph. So, the factor graph depends on the factorization of the function $g$.

We collect some simple properties of factor graphs.
Lemma 2.2.4. Considering factor graphs as given in Definition 2.2.2, the following properties hold.
(i) A factor graph is a bipartite graph.
(ii) For any simple bipartite graph $G=\left(V_{1}, V_{2}, E\right)$ of vertices $V=V_{1} \dot{\cup} V_{2}$ and edges $E \subseteq V_{1} \times V_{2}$, we can equip $V_{1}$ with variables and $V_{2}$ with factors such that $G$ is a factor graph.
(iii) For a function $g: D \rightarrow R$ with a factorization $\prod_{j \in J} f_{j}\left(X_{j}\right)$ as in (2.13) there exists exactly one factor graph $\mathcal{F}$ representing this factorization.

Proof. (i) Follows directly from the split of the set of vertices in variable nodes and factor nodes and that no two factor nodes and no two variable nodes are connected in the graph.
(ii) For each node in $V_{v}:=V_{1}$, we introduce a variable $x_{i}$ together with a set of values $A_{i}$ as the domain of this variable and for each node in $V_{f}:=V_{2}$, we introduce a function $f_{j}$ with a semiring $R$ as common codomain. Let the function $f_{j}$ depend on all variables, the respective node is connected to. The product of all these functions gives a function $g\left(x_{1}, \ldots, x_{n}\right)$ with domain $A_{1} \times \cdots \times A_{n}$ and codomain $R$. Thus, we have created a function $g$ with a factorization according to Definition 2.2.2 that has the bipartite graph $G$ as its factor graph.
(iii) This can be seen by introducing variable nodes for each variable $x_{i}$ of $g$ and factor nodes for each factor $f_{j}$ of $g$ and connecting them accordingly.

So we have seen, how we can turn functions into factor graphs and factor graphs into functions. This allows us to switch between factor graphs and functions easily.

### 2.2.1. Applications of Factor Graphs

Factor graphs offer a variety of applications. The most important one in our setting is probabilistic modeling with factor graphs, used for example in certain navigation and location tasks. Moreover, factor graphs are used for instance in robotics [10], coding theory [42], and artificial intelligence [63].

In the pose estimation problem of Section 3.2, we search for an assignment of the different parts of a truck to poses, that fits some prior information about relations between the parts and the observations of the parts best. We can translate this in probability theoretical terms: We search for the state $X$ (a variable assignment) that is most likely under the given preconditions, assumptions, and observations $Z$, thus, we want to maximize the posterior density $p(X \mid Z)$ [13, Section 1.6]. The following lemma can rephrase this maximization problem.

Lemma 2.2.5. The maximum a posteriori estimate $X^{\mathrm{MAP}}:=\arg \max _{X} p(X \mid Z)$ is given by the joint probability function $p(X, Z)$ as

$$
X^{\mathrm{MAP}}=\underset{X}{\arg \max } p(X, Z)
$$

Proof. Bayes' law states in this context that

$$
p(X \mid Z)=\frac{p(Z \mid X) p(X)}{p(Z)}
$$

The preconditions, assumptions, and observations $Z$ are given, hence the term $p(Z)$ is some constant, positive factor, not influencing the maximal argument. Therefore, maximizing the posterior $p(X \mid Z)$ translates as

$$
\underset{X}{\arg \max } p(X \mid Z)=\underset{X}{\arg \max } \frac{p(Z \mid X) p(X)}{p(Z)}=\underset{X}{\arg \max } p(Z \mid X) p(X) .
$$

The term $p(Z \mid X) p(X)$ equals the joint probability $p(X, Z)$ according to the definition of conditional probability [6, Section 1.3]. Thus, the maximum a posteriori estimate $X^{\text {MAP }}$ can be computed by maximizing the joint probability $p(X, Z)$.

Remark 2.2.6. The term $p(Z \mid X)$ is also called likelihood (function) in statistics and can be denoted with $L(X \mid Z)$, indicating, that this is seen as a function of $X$ and not as a function of $Z$ [6, Section 6.3.1].


Figure 2.3.: Factor graph modeling a sensor observing landmarks over time
Example 2.2.7. Figure 2.3 shows a simple example of a factor graph $\mathcal{F}$, similar to [13, Chapter 1], modeling the situation of a sensor $x$ moving past some landmarks $l_{1}, l_{2}, l_{3}$ of unknown poses, e.g. the three wheels on one side of a truck. The set of variable nodes $V$ of the factor graph $\mathcal{F}=(V, F, E)$ is given by $V=\left\{x_{1}, x_{2}, x_{3}, l_{1}, l_{2}, l_{3}\right\}$ and the set $F$ of the factor nodes is visualized by the black squares. The variables $x_{i}$ for $i=1,2,3$ denote the sensor positions at three consecutive time steps. If the sensor detects at time step $i$ the landmark $l_{j}$, the observation $z_{l_{j}}^{(i)}$ is generated. These observations are seen as fixed given values, so there are no variable nodes for them. A factor graph is undirected, the arrows in the factor graph indicate that the poses of the landmarks and the sensor influence the observations, resulting in the probability densities $p\left(z_{l_{j}}^{(i)} \mid x_{i}, l_{j}\right)$.

With the definition of conditional probability [6, Section 1.3], the joint probability $p\left(x_{1}, x_{2}, x_{3}\right)$ is given by

$$
p\left(x_{1}, x_{2}, x_{3}\right)=p\left(x_{3} \mid x_{2}, x_{1}\right) p\left(x_{2} \mid x_{1}\right) p\left(x_{1}\right)=p\left(x_{3} \mid x_{2}\right) p\left(x_{2} \mid x_{1}\right) p\left(x_{1}\right)
$$

where the last equality follows if we assume that the Markov property holds. The Markov property states in this case that the pose of the sensor at time step 3 is just dependent
on the pose of the sensor at time step 2 [6, Section 5.8.5]. Similarly, the joint probability $p(X, Z)$ of all sensor poses, landmarks, and observations is given by

$$
p(X, Z)=p\left(x_{3} \mid x_{2}\right) p\left(x_{2} \mid x_{1}\right) p\left(x_{1}\right) p\left(l_{1}\right) p\left(l_{2}\right) p\left(l_{3}\right) \prod_{i, j} p\left(z_{l_{j}}^{(i)} \mid x_{i}, l_{j}\right)
$$

which is exactly the global function $f(X)$ defined as the product of all factors of the factor graph $\mathcal{F}$.

A common application of factor graphs among many other applications in robotics is Simultaneous Localization and Mapping (SLAM) [13, Chapter 2]. It is similar to our pose estimation approach and the scene displayed in Figure 2.3 is an instance of a SLAM problem. A robot tries to orient itself in an unknown environment. Equipped with some sensors, the robot moves around and detects objects (here called landmarks) and their approximate relative pose to the robot. These measurements are represented by factor nodes between the variables $x_{t}$ of the robot's pose at a time step $t$ and variables $l_{i}$ introduced for each landmark. Furthermore, odometry measurements - information about the movement (e.g. velocity and acceleration) of the robot [56, Section 5.4] - are taken into consideration. They translate to factors between the variables ( $x_{1}, x_{2}$, and $x_{3}$ in the example above) denoting the position of the robot at certain time steps. Prior factors on landmarks or the robot's starting pose enable absolute location in the environment.

Factor graphs can be used in coding theory to model and decode certain codes [42]. The indicator function $I_{C}: A^{n} \rightarrow\{0,1\}$ for a code $C$ over the alphabet $A$, that maps a word $c \in A^{n}$ to 1 if $c$ is a code word and to 0 otherwise, is interpreted as the global function of a factor graph. This factor graph allows efficient decoding, for instance with the sum-product algorithm.

In artificial intelligence, factor graphs can be used to model neural networks. Zhang et al. describe in [63] factor graph neural networks to model dependencies between different variables.

### 2.3. Optimization

Optimization is the task of minimizing or maximizing a function $f$ under certain constraints [25]. Finding a minimum of the function $g(x)=x^{2}-3 x+2$ can be done by exploiting the properties of continuously differentiable functions. In many real-life scenarios, there is no explicit representation of a function that can easily be differentiated. Hence, we need optimization methods to get as close as possible to a local or global optimum.

In Section 2.3.1, an optimization technique is introduced. Levenberg-Marquardt optimization is used by GTSAM [11], the Python and C++ package employed in the solution of the pose estimation problem, to optimize factor graphs.

Many optimization problems are set in a vector space, e.g. in $\mathbb{R}^{n}$, where simple and intuitive optimization techniques are applicable. However, there are problems where a function $f$ needs to be optimized on a (nonlinear) manifold $M$. There, we have to consider the issue of moving on the manifold efficiently, which can be solved by retractions as described in Section 2.3.2 and applied to $\mathrm{SO}_{2}$ and $\mathrm{SO}_{3}$ in Section 2.3.3. Section 2.3.4 deals with optimization in $\mathrm{SE}_{3}$.

Furthermore, factor graphs can be optimized as well if they are equipped with some probability structure. Section 2.3.5 deals with the optimization of factor graphs.

### 2.3.1. Levenberg-Marquardt Optimization

In Section 2.1, we have established the necessary geometric background for optimizing the pose of an object. Now, we will briefly discuss the actual scheme for nonlinear optimization used in this project. In particular, in Section 3.2 we describe the part of the project that deals with pose estimation with factor graphs using the Python and C++ library GTSAM [11] which provides implementations of various optimization algorithms. We use the Levenberg-Marquardt algorithm [45] that can be seen as a combination of gradient descent and Gauss-Newton optimization [13, Section 2.5.3]. Marquardt describes this algorithm and the theoretical background in [45].

Gradient descent, Gauss-Newton, and Levenberg-Marquardt optimization are iterative algorithms to optimize (w.l.o.g. minimize) a function $g$, starting from an initial estimate $x^{(0)}$ and updating an estimate $x^{(t)}$ by the rule

$$
\begin{equation*}
x^{(t+1)}=x^{(t)}+\alpha \delta^{(t)} \tag{2.14}
\end{equation*}
$$

with $\alpha \in \mathbb{R}$ and some update step $\delta^{(t)}$ depending on the method. The scaling factor $\alpha$ is chosen concerning the specific use case, ensuring that the convergence speed is fast enough on the one hand and that the updates are safe, i.e. the steps are not too big to risk divergence, on the other hand. This process continues until the solutions $x^{(t)}$ converge, i.e. $\delta^{(t)}$ is smaller than some threshold. The following considerations are based on [13, Section 2.5] where these three methods are discussed.

Gradient descent is a simple nonlinear optimization technique to minimize a differentiable function $g$ by taking steps in the direction of the steepest descent, given by the negative gradient $-\operatorname{grad} g\left(x^{(t)}\right)$ of the current guess $x^{(t)}$. Thus, the update rule for gradient descent is given by

$$
x^{(t+1)}=x^{(t)}+\alpha \delta_{\mathrm{GD}}^{(t)}=x^{(t)}-\alpha \operatorname{grad} g\left(x^{(t)}\right)
$$

Gradient descent has a slow convergence speed close to the minimum.
The Gauss-Newton method is a technique to minimize a sum of squared continuously differentiable functions. This least squares problem is given by

$$
\begin{equation*}
\min _{x \in \mathbb{R}^{n}} \sum_{i=1}^{m}\left(g_{i}(x)\right)^{2} \tag{2.15}
\end{equation*}
$$

with the function $g=\left(g_{1}, \ldots, g_{m}\right)$ and $g_{i}: \mathbb{R}^{n} \rightarrow \mathbb{R}$ for each $i=1, \ldots, m$. The idea is to approximate $g$ by a Taylor series of first order [45], thus

$$
\begin{equation*}
g(x) \approx g\left(x_{0}\right)+\sum_{j=1}^{n} \operatorname{grad}_{j} g\left(x_{0}\right)\left(x-x_{0}\right)_{j}=g\left(x_{0}\right)+J\left(x_{0}\right)\left(x-x_{0}\right) \tag{2.16}
\end{equation*}
$$

where $J$ denotes the well-known Jacobian matrix

$$
J(x):=\left(\begin{array}{ccc}
\frac{\partial g_{1}}{\partial x_{1}}(x) & \cdots & \frac{\partial g_{1}}{\partial x_{n}}(x) \\
\vdots & \ddots & \vdots \\
\frac{\partial g_{m}}{\partial x_{1}}(x) & \cdots & \frac{\partial g_{m}}{\partial x_{n}}(x)
\end{array}\right)
$$

Minimizing as in (2.15) with this linearization of $g$ instead of $g$ itself can be done by taking the gradient of the squared norm of the right-hand side of (2.16) and setting it to zero. This results in the equation

$$
\operatorname{grad}\left\|g\left(x_{0}\right)+J\left(x_{0}\right)\left(x-x_{0}\right)\right\|^{2}=0 \quad \Rightarrow \quad J^{T}\left(x_{0}\right)\left(J\left(x_{0}\right)\left(x-x_{0}\right)+g\left(x_{0}\right)\right)=0 .
$$

Here, we take the linearization point $x^{(t)}$ and evaluate the Taylor series expansion at the point $x^{(t+1)}=x^{(t)}+\delta_{\mathrm{GN}}^{(t)}$ for some unknown $\delta_{\mathrm{GN}}^{(t)}$. Therefore, the update step $\delta_{\mathrm{GN}}^{(t)}$ is implicitly defined by the equation

$$
\begin{equation*}
J^{T}\left(x^{(t)}\right) J\left(x^{(t)}\right) \delta_{\mathrm{GN}}^{(t)}=-J^{T}\left(x^{(t)}\right) g\left(x^{(t)}\right) \tag{2.17}
\end{equation*}
$$

Thus, the Gauss-Newton update rule is

$$
x^{(t+1)}=x^{(t)}-\alpha \underbrace{\left(J^{T}\left(x^{(t)}\right) J\left(x^{(t)}\right)\right)^{-1} J^{T}\left(x^{(t)}\right) g\left(x^{(t)}\right)}_{=\delta_{\text {GN }}^{(t)}},
$$

where $\alpha$ is again some scaling factor [45]. This method can have poor convergence behavior if $g$ is not nearly quadratic [13, Section 2.5.2].

In the Levenberg-Marquardt optimization, the Gauss-Newton update is modified by introducing a real factor $\lambda \geq 0$ in (2.17) to get the equation

$$
\begin{equation*}
\left(J^{T}\left(x^{(t)}\right) J\left(x^{(t)}\right)+\lambda \operatorname{diag}\left(J^{T}\left(x^{(t)}\right) J\left(x^{(t)}\right)\right)\right) \delta_{\mathrm{LM}}^{(t)}=-J^{T}\left(x^{(t)}\right) g\left(x^{(t)}\right) . \tag{2.18}
\end{equation*}
$$

where $\operatorname{diag}(A)$ denotes the diagonal matrix $\operatorname{diag}\left(a_{11}, a_{22}, \ldots, a_{l l}\right)$ consisting of the entries of the diagonal of a quadratic matrix $A \in \mathbb{R}^{l \times l}$. This results in larger steps towards the direction of the steepest descent if the gradient is small. If the gradient is big, the steps are smaller to reduce the risk of divergence. Another modification can be made by rejecting steps that lead to an increase in the value that should be minimized. If a step is rejected, the value of $\lambda$ is increased (e.g. in [13, Algorithm 2.1] $\lambda$ is multiplied by 10) and the last step is retaken with the new $\lambda$. If a step is accepted, $\lambda$ is diminished again (e.g. $\lambda$ is divided by 10). Marquardt suggests in [45] that this algorithm combines the advantage of gradient descent that it converges from rather far away, and the advantage of the Gauss-Newton method that it converges rapidly when we are already close to a solution.

### 2.3.2. Optimization on Manifolds

Optimization methods like gradient descent, Gauss-Newton, and Levenberg-Marquardt rely on the update rule (2.14) to gradually improve some estimate $x^{(t)}$. This works well in vector spaces since $\delta^{(t)}$ is rather easy to define and compute. On a manifold $M$, it is not that easy to take a step and still stay on the manifold, thus, resulting in a new valid estimate $x^{(t+1)} \in M$. Absil, Mahony, and Sepulchre describe in [1, Section 4.1] a method to take a step in a vector space and consequently bring the resulting point back onto the manifold. A function that achieves this mapping from a vector space back to the manifold is called a retraction.

In a topological sense, a retraction $r$ is a continuous function from a topological space $X$ to a subspace $Y$ of $X$ with $r(y)=y$ for all $y \in Y$ [31, Chapter 0]. Clearly, $r$ is idempotent, i.e. $r \circ r=r$, and therefore the topological analog to a projection in linear algebra. Here, we need retractions on manifolds. The idea stays the same: A retraction brings a point from the tangent space onto the manifold. The formal definition of retractions on manifolds given in [1, Definition 4.1.1] is as follows.

Definition 2.3.1. Let $M$ be a manifold. A retraction on $M$ is a smooth function

$$
\mathcal{R}: T M \rightarrow M
$$

such that the following properties hold for the restriction $\mathcal{R}_{x}:=\left.\mathcal{R}\right|_{T_{x} M}$ for every $x \in M$ (i.e. in $\mathcal{R}_{x}$, we drop the first part of a pair $\left.(x, \xi) \in T M\right)$.
(i) Let $0_{x}$ denote the zero vector of the vector space $T_{x} M$, then $\mathcal{R}_{x}\left(0_{x}\right)=x$.
(ii) The differential $D \mathcal{R}_{x}\left(0_{x}\right)[\cdot]$ is equal to the identity $\operatorname{id}_{T_{x} M}$ on the tangent space $T_{x} M$.

These conditions ensure that the gradient at $x$ is preserved under $\mathcal{R}_{x}$ [1, Section 4.1]. This can be visualized as in Figure 2.4.


Figure 2.4.: Visualization of a retraction taken from [1, Figure 4.1].
A deeper examination of differential geometry and the theory behind Definition 2.3.1 would be beyond the scope of this work. We will just make a brief remark and provide references to the corresponding sources.
Remark 2.3.2. The zero element $0_{x}$ of $T_{x} M$ in condition (i) of Definition 2.3.1 is according to (2.5) the pre-image of $0_{n} \in \mathbb{R}^{n}$ under some $\vartheta_{\varphi}^{x, M}$ of Lemma 2.1.23. Thus, the tangent vector $0_{x}$ belongs to some (and therefore all) curve(s) $\gamma$ with $(\varphi \circ \gamma)^{\prime}(0)=0_{n}$.

Part (ii) of Definition 2.3.1 is often referred to as the local rigidity condition [1, Section 4.1]. The differential $D \mathcal{R}_{x}\left(0_{x}\right)[\cdot]$ is a function that takes a tangent vector $\eta$ of the domain of $\mathcal{R}_{x}$ at $0_{x}$ and maps it to the tangent vector $D \mathcal{R}_{x}\left(0_{x}\right)[\eta]$ in the tangent space of the codomain of $\mathcal{R}_{x}$. So to be precise, $D \mathcal{R}_{x}\left(0_{x}\right)[\cdot]$ is a function $T_{0_{x}}\left(T_{x} M\right) \rightarrow T_{x} M$ and is only equal to the identity on $T_{x} M$ if we identify $T_{0_{x}}\left(T_{x} M\right)$ and $T_{x} M$ as described in [1, Section 3.5.2]. See [1, Section 3.5.6] for a more detailed description of this differential.

Using a retraction $\mathcal{R}$ in an optimization problem on a manifold $M$ offers two advantages [1, Section 4.1]. First, $\mathcal{R}$ brings elements of tangent spaces $T_{x} M$ back onto the manifold $M$. So starting from an $x^{(t)} \in M$ we can easily take steps in the tangent space $T_{x^{(t)}} M$ since it is a vector space and, furthermore, closely related to $\mathbb{R}^{n}$ due to the map defined in Lemma 2.1.23. The retraction $\mathcal{R}$ maps this new point in $T_{x^{(t)}} M$ onto the manifold $M$ and therefore delivers a new estimate $x^{(t+1)}$.

Second, in an optimization problem, there is usually a cost function $c: M \rightarrow \mathbb{R}$ on the manifold $M$ that should be minimized. But when the steps are taken in the tangent space, the cost function needs to be lifted to $T_{x} M$ as well. The function

$$
c^{\mathcal{R}}:=c \circ \mathcal{R}: T M \rightarrow \mathbb{R}
$$

lifts the cost function to the tangent bundle. For a point $x \in M$, we denote the restriction of $c^{\mathcal{R}}$ to the tangent space $T_{x} M$ by

$$
c_{x}^{\mathcal{R}}:=\left.c^{\mathcal{R}}\right|_{T_{x} M}=c \circ \mathcal{R}_{x}: \quad T_{x} M \rightarrow \mathbb{R}
$$

which is a function from a vector space to $\mathbb{R}$. Due to the chain rule $[1$, Section 1.3$]$ and condition (ii) of Definition 2.3.1, the differential of $c_{x}^{\mathcal{R}}$ computes as $D c_{x}^{\mathcal{R}}\left(0_{x}\right)=D c(x)$.
Remark 2.3.3. Absil, Mahony, and Sepulchre state in [1, (4.4)] that even

$$
\operatorname{grad} c_{x}^{\mathcal{R}}\left(0_{x}\right)=\operatorname{grad} c(x)
$$

holds if the manifold $M$ is endowed with a Riemannian metric (which is the case for $\mathrm{SO}_{n}$ ). This would again go beyond the scope of this work. See [1, Section 3.6] for more information about Riemannian metrics.
Example 2.3.4. There are several retractions for the special orthogonal group $\mathrm{SO}_{n}$. We will focus on a retraction that uses the exponential map according to [2, Example 2]. For a rotation $R \in \mathrm{SO}_{n}$ a retraction is given by

$$
\begin{equation*}
\mathcal{R}_{R}: T_{R} \mathrm{SO}_{n} \rightarrow \mathrm{SO}_{n}: \eta \mapsto R e^{R^{-1} \eta} . \tag{2.19}
\end{equation*}
$$

The tangent space $T_{R} \mathrm{SO}_{n}$ is given by (2.9), so $\eta$ is of the form $R \hat{\xi}$ for some $\hat{\xi} \in$ Skew $_{n}$ (the hat-operator is used for consistent notation with the next Chapter, see Definition 2.3.5). Hence, $\mathcal{R}_{R}(\eta)$ reduces to

$$
\mathcal{R}_{R}(\eta)=R e^{R^{-1} \eta}=R e^{R^{-1} R \hat{\xi}}=R e^{\hat{\xi}}
$$

The exponential $e^{\hat{\xi}}$ of the skew-symmetric matrix $\hat{\xi}$ is according to Theorem 2.1.37 in $\mathrm{SO}_{n}$. Thus, the matrix product $R e^{\hat{\xi}}$ is in $\mathrm{SO}_{n}$ and the function $\mathcal{R}_{R}$ well-defined.

To get a retraction in the sense of Definition 2.3.1, let $\mathcal{R}$ be the function that maps a pair $(R, \eta) \in T \mathrm{SO}_{n}$ to $\mathcal{R}_{R}(\eta)$. Condition (i) holds due to the computation

$$
\mathcal{R}_{R}\left(0_{R}\right)=R e^{R^{-1} 0_{R}}=R e^{R^{-1} R 0_{n}}=R e^{0_{n}}=R I_{n}=R
$$

with the zero element $0_{R}$ of the tangent space $T_{R} \mathrm{SO}_{n}$ and the zero element $0_{n}$ of the vector space $\mathrm{Skew}_{n}$. For the second part of the definition, we have to compute the differential $D \mathcal{R}\left(0_{R}\right)[\cdot]$. Najfeld and Havel give in [48, Section 1.2] the general definition

$$
D e^{t A}[V]=\lim _{h \rightarrow 0} \frac{1}{h}\left(e^{t(A+h V)}-e^{t A}\right)
$$

and an explicit way to calculate the directional derivative of the matrix exponential $e^{t A}$ in the direction $V$ by

$$
D e^{t A}[V]=\int_{0}^{t} e^{(t-\tau) A} V e^{\tau A} d \tau
$$

Here, with $\eta=R \hat{\xi}_{\eta} \in T_{R} \mathrm{SO}_{n}$ and $\zeta=R \hat{\xi}_{\zeta} \in T_{R} \mathrm{SO}_{n}$, this translates to

$$
\begin{aligned}
D \mathcal{R}_{R}\left(0_{R}\right)[\zeta] & =\left.\lim _{h \rightarrow 0} \frac{1}{h}\left(R e^{R^{-1}(\eta+h \zeta)}-R e^{R^{-1} \eta}\right)\right|_{\eta=0_{R}} \\
& =\left.\lim _{h \rightarrow 0} \frac{1}{h}\left(R e^{R^{-1}\left(R \hat{\xi}_{\eta}+h R \hat{\xi}_{\zeta}\right)}-R e^{R^{-1} R \hat{\xi}_{\eta}}\right)\right|_{R \hat{\xi}_{\eta}=0_{n}} \\
& =\left.R \lim _{h \rightarrow 0} \frac{1}{h}\left(e^{\hat{\xi}_{\eta}+h \hat{\xi}_{\zeta}}-e^{\hat{\xi}_{\eta}}\right)\right|_{\hat{\xi}_{\eta}=0_{n}} \\
& =R D e^{0_{n}}\left[\hat{\xi}_{\zeta}\right] \\
& =R \int_{0}^{1} e^{(1-\tau) 0_{n}} \hat{\xi}_{\zeta} e^{\tau 0_{n}} d \tau \\
& =R \int_{0}^{1} I_{n} \hat{\xi}_{\zeta} I_{n} d \tau=R \hat{\xi}_{\zeta}=\zeta .
\end{aligned}
$$

This shows that $D \mathcal{R}_{R}\left(0_{R}\right)[\cdot]$ is the identity on $T_{R} \mathrm{SO}_{n}$. Thus, $\mathcal{R}$ is a retraction.

### 2.3.3. Optimization on $\mathrm{SO}_{2}$ and $\mathrm{SO}_{3}$

After the preparations made in Section 2.1, we can formalize small steps on a rotation manifold. Instead of adding an arbitrary, small matrix, we take small rotations in some natural representation, translate that into a matrix, use the exponential map for quadratic matrices as a retraction to get a rotation matrix, and then compose it with the base rotation.

In $\mathrm{SO}_{2}$, this process is quite straightforward as described in [13, Section 6.1.5]. The special orthogonal group $\mathrm{SO}_{2}$ is according to Theorem 2.1.28 a $\left(\frac{2(2-1)}{2}=1\right)$-dimensional manifold. The tangent space $T_{B} \mathrm{SO}_{2}$ at some point $B \in \mathrm{SO}_{2}$, given by (2.9), is a 1dimensional vector space. A planar rotation around the origin can be uniquely defined by a single number $\xi \in \mathbb{R}$. In [13, Section 6.1.3], Dellaert and Kaess refer to this number and its $\mathrm{SO}_{3}$-equivalent (see below), that represents a step in an incremental rotation, in this context as local coordinates or a local parametrization. Here, we follow their proposed way of optimizing on the rotation manifolds $\mathrm{SO}_{2}$ and $\mathrm{SO}_{3}$.

The special orthogonal group $\mathrm{SO}_{3}$ is a $\left(\frac{3(3-1)}{2}=3\right)$-dimensional manifold with the tangent space $T_{B} \mathrm{SO}_{3}=\left\{B S \mid S \in\right.$ Skew $\left._{3}\right\}$. As seen in Section 2.1.1, a rotation in $\mathbb{R}^{3}$ can be represented by an axis $a \in S^{2}$ and an angle $\alpha \in \mathbb{R}$. Since $\|a\|=1$, the rotation is also uniquely defined by the local coordinates $\xi:=\alpha a \in \mathbb{R}^{3}$.

To use the retraction $\mathcal{R}$ given in example 2.3.4 for $\mathrm{SO}_{2}$ and $\mathrm{SO}_{3}$, we need tangent vectors of these groups, i.e. skew-symmetric matrices, instead of the local coordinates stated above. In [43, Section 3], this lifting from local coordinates in $\mathbb{R}^{3}$ to skew-symmetric matrices is described by the matrix $\hat{\xi} \in$ Skew $_{3}$ that is the cross-product matrix of $\xi$, i.e. $\hat{\xi} c=\xi \times c$ for any $c \in \mathbb{R}^{3}$. The $\mathbb{R}^{2}$ equivalent is the function that maps a vector $c=\left(c_{1}, c_{2}\right)^{T}$ to its orthogonal vector $\left(-c_{2}, c_{1}\right)^{T}$. The matrix of this linear mapping is again skew-symmetric. The hat-operator defines this mapping from local coordinates to cross-product matrices.

Definition 2.3.5. The hat-operator for planar rotations is the function defined as

$$
: \mathbb{R} \rightarrow \mathbb{R}^{2 \times 2}: \quad \xi \mapsto \hat{\xi}:=\left(\begin{array}{rr}
0 & -\xi \\
\xi & 0
\end{array}\right) .
$$

The hat-operator for rotations in $\mathbb{R}^{3}$ is the function defined as

$$
\vdots \mathbb{R}^{3} \rightarrow \mathbb{R}^{3 \times 3}: \quad \xi=\left(\begin{array}{l}
\xi_{x} \\
\xi_{y} \\
\xi_{z}
\end{array}\right) \mapsto \hat{\xi}:=\left(\begin{array}{rrr}
0 & -\xi_{z} & \xi_{y} \\
\xi_{z} & 0 & -\xi_{x} \\
-\xi_{y} & \xi_{x} & 0
\end{array}\right) .
$$



Figure 2.5.: Visualization of $\vec{\omega}$ and $\vec{v}$ as in [55, Figure 7-4].
We provide a physical interpretation of why it is reasonable to use the hat-operator here. The velocity vector $\vec{v}$ of a point that rotates around an axis is given by the cross product $\vec{v}=\vec{\omega} \times \vec{s}$ of the angular velocity $\vec{\omega}$ and the position $\vec{s}$ of the object [38, Chapter VI $\S 31]$. This is visualized in Figure 2.5. The angular velocity $\vec{\omega}$ refers to the local coordinates $\xi \in \mathbb{R}^{3}$ defined by the product $\alpha a$ of the angle $\alpha$ and the axis $a$. The tangent vectors in the tangent space $T_{B} \mathrm{SO}_{3}$ can be seen as the speed vectors of any curves at the point $B[40$, Section 3.5]. Thus, the velocity vector $\vec{v}$ refers to the tangent vectors in $T_{B} \mathrm{SO}_{3}$ given by the product of $B$ with $\hat{\xi}$.

So, when optimizing in $\mathrm{SO}_{2}$, the hat-operator transforms the rotation angle $\xi$ into a skew-symmetric $(2 \times 2)$ matrix $\hat{\xi}$. Starting at a rotation $R_{\theta}$, defined by the angle $\theta$, the
retraction can be used to update this rotation by a local parametrization $\xi$ as

$$
\begin{aligned}
\mathcal{R}_{R_{\theta}}\left(R_{\theta} \hat{\xi}\right) & =R_{\theta} e^{R_{\theta}^{-1} R_{\theta} \hat{\xi}}=R_{\theta} e^{\hat{\xi}} \\
& \stackrel{2.1 .36}{=}\left(\begin{array}{rr}
\cos \theta & -\sin \theta \\
\sin \theta & \cos \theta
\end{array}\right)\left(\begin{array}{rr}
\cos \xi & -\sin \xi \\
\sin \xi & \cos \xi
\end{array}\right) \\
& \stackrel{2.1 .9}{=}\left(\begin{array}{rr}
\cos (\theta+\xi) & -\sin (\theta+\xi) \\
\sin (\theta+\xi) & \cos (\theta+\xi)
\end{array}\right)
\end{aligned}
$$

We introduce a notation for updating a base rotation $R_{0}$ by local coordinates $\xi$.
Definition 2.3.6. Given a rotation $R_{0}$ in $\mathrm{SO}_{2}$ or $\mathrm{SO}_{3}$ and local coordinates $\xi$ in $\mathbb{R}$ respectively $\mathbb{R}^{3}$ as described above, we can define the local update $\oplus$ of the rotation $R_{0}$ by the local coordinates $\xi$ as

$$
R_{0} \oplus \xi:=\mathcal{R}_{R_{0}}\left(R_{0} \hat{\xi}\right)
$$

Note, that the matrix $e^{\hat{\xi}}$ for $\xi=\alpha a$, that appears in a local update in $\mathrm{SO}_{3}$, has the explicit representation

$$
e^{\hat{\xi}}=I_{3}+\frac{\sin \alpha}{\alpha} \hat{\xi}+\frac{1-\cos \alpha}{\alpha^{2}} \hat{\xi}^{2}
$$

according to Rodrigues' formula 2.1.38 and since

$$
\|\xi\|=\sqrt{\left(\alpha a_{x}\right)^{2}+\left(\alpha a_{y}\right)^{2}+\left(\alpha a_{z}\right)^{2}}=|\alpha|\|a\|=|\alpha|
$$

holds. Hence, this retraction can be computed efficiently in $\mathrm{SO}_{2}$ and $\mathrm{SO}_{3}$.
Example 2.3.7. Let us examine how a simplified version of the pose estimation problem of Section 3.2 would translate to an optimization problem on $\mathrm{SO}_{3}$. For now, we just want to optimize for the orientation, so assume there is a camera on a pole at a fixed, known position on one side of a truck. The camera can only rotate itself but cannot change its position. It can detect the three wheels and the loading edge on the side of the truck (the view of the camera might look like in Figure 3.12). Each detection generates a vector $z=\left(z^{(e)}, z^{\left(w_{1}\right)}, z^{\left(w_{2}\right)}, z^{\left(w_{3}\right)}\right)^{T} \in \mathbb{R}^{12}$ where $z^{(e)}, z^{\left(w_{1}\right)}, z^{\left(w_{2}\right)}, z^{\left(w_{3}\right)} \in \mathbb{R}^{3}$ denote the measurements of the loading edge and the three wheels as the direction vectors from the camera to the respective parts of the truck (simplified as points) in the current camera frame (i.e. the $x$-axis points in the direction the camera is looking at, the $z$-axis points upwards, and the $y$-axis points to the left, such that all axes together create a right-handed coordinate system).

Assume that a rough model of the relations between the different parts of the truck is known and that we have access to an estimation function

$$
h: \mathrm{SO}_{3} \rightarrow \mathbb{R}^{12}
$$

For a given rotation $R \in \mathrm{SO}_{3}$, this function $h$ estimates the corresponding measurement $z_{R} \in \mathbb{R}^{12}$ based on the assumed model of the truck. This function is not surjective, in particular, it does not have an inverse function. Furthermore, $h$ might be way too complicated to invert it even if the codomain were restricted to $h\left(\mathrm{SO}_{3}\right)$ and $h$ were injective. On top of that, the measurements obtained by the camera are not exact but noisy and the
model of the truck is not exact as well. Thus, a measurement obtained by the camera does not have to be an element of $h\left(\mathrm{SO}_{3}\right)$. For these various reasons, getting the rotation of the camera from a given measurement $z \in \mathbb{R}^{12}$ is a challenging task that can be tackled by optimizing

$$
R^{*}=\underset{R \in \mathrm{SO}_{3}}{\arg \min }\|h(R)-z\|^{2}
$$

In the following, we will examine, how $\|h(R)-z\|^{2}$ can be optimized similar to [13, Section 6.1.3] with Levenberg-Marquardt optimization discussed in Section 2.3.1.

Given an estimation function $h: \mathrm{SO}_{3} \rightarrow \mathbb{R}^{n}$, the goal is to find

$$
R^{*}=\underset{R \in \mathrm{SO}_{3}}{\arg \min }\|h(R)-z\|^{2}
$$

Methods like gradient descent and Levenberg-Marquardt start at some initial estimate $R^{(0)}$ and iteratively take steps $\delta^{(t)}$ towards a new estimate $R^{(t+1)}$ for $t \geq 0$ by minimizing

$$
\delta^{(t)}=\underset{\xi \in \mathbb{R}^{3}}{\arg \min }\left\|h\left(R^{(t)} \oplus \xi\right)-z\right\|^{2}
$$

to get $R^{(t+1)}=R^{(t)} \oplus \delta^{(t)}$. Let's define $g_{R}(\xi):=h(R \oplus \xi)$. An approximation of the Jacobian $G_{R}$ of $g_{R}$ can be computed through, for instance, numerical differentiation or automated differentiation. As in (2.16), we can approximate $g_{R}(\xi)$ by

$$
g_{R}(\xi) \approx g_{R}(0)+G_{R}(0)(\xi-0)=h(R)+G_{R} \xi
$$

Therefore, we can define the update step $\delta_{\mathrm{LM}}^{(t)}$ as in (2.18).

### 2.3.4. Optimization in $\mathrm{SE}_{3}$

Until now, we just considered optimization of rotations. In the pose estimation problem of Section 3.2, we search for optimal poses defined by elements of $\mathrm{SE}_{3}$ rather than just optimal rotations. This is just a simple generalization from $\mathrm{SO}_{3}$ to $\mathrm{SE}_{3}$, analogously to [13, Sections 6.2.2 and 6.2.3].

Local coordinates $\xi$ in $\mathrm{SO}_{3}$ were given by the product of an axis $a \in S^{2}$ and an angle $\alpha \in \mathbb{R}$. For local coordinates in $\mathrm{SE}_{3}$, we simply expand the local coordinates of $\mathrm{SO}_{3}$ by a vector $v \in \mathbb{R}^{3}$ modeling translations in $\mathbb{R}^{3}$. Thus, we receive 6 -dimensional local coordinates $\xi$ of the 6-dimensional manifold $\mathrm{SE}_{3}$ (see Theorem 2.1.29).

The tangent space $T_{(R, t)} \mathrm{SE}_{3}$ of the product manifold $\mathrm{SE}_{3}=\mathrm{SO}_{3} \times \mathbb{R}^{3}$ is according to [59, solution to Problem 8.7] isomorphic to the product of the tangent spaces $T_{R} \mathrm{SO}_{3} \times T_{t} \mathbb{R}^{3}$. Thus, we view tangent vectors to $\mathrm{SE}_{3}$ as a pair $(\eta, s) \in T_{R} \mathrm{SO}_{3} \times T_{t} \mathbb{R}^{3}$ or a matrix

$$
\left(\begin{array}{ll}
\eta & s \\
0 & 0
\end{array}\right) \in \mathbb{R}^{4 \times 4}
$$

if necessary, with $\eta \in R$ Skew $_{3}$ according to (2.9) and $s \in \mathbb{R}^{3}$.

Therefore, we can define a retraction for the special Euclidean group $\mathrm{SE}_{3}$ as in Example 2.3.4. For $(R, t) \in \mathrm{SE}_{3}$ and $(\eta, s)=\left(R \hat{\xi}_{\mathrm{SO}_{3}}, s\right) \in T_{R} \mathrm{SO}_{3} \times \mathbb{R}^{3}$, we define the retraction as

$$
\mathcal{R}_{(R, t)}: T_{(R, t)} \mathrm{SE}_{3} \rightarrow \mathrm{SE}_{3}:(\eta, s) \mapsto\left(\begin{array}{cc}
R & t  \tag{2.20}\\
0 & 1
\end{array}\right)\left(\begin{array}{cc}
e^{R^{-1} \eta} & R^{-1} s \\
0 & 1
\end{array}\right)=\left(\begin{array}{cc}
R e^{R^{-1} \eta} & s+t \\
0 & 1
\end{array}\right) .
$$

We see that $\mathcal{R}_{(R, t)}(\eta, s)=\left(\mathcal{R}_{R}^{\mathrm{SO}_{3}}(\eta), s+t\right) \in \mathrm{SE}_{3}$ with the retraction $\mathcal{R}_{R}^{\mathrm{SO}_{3}}$ of Example 2.3.4. At the zero element $0_{(R, t)}$ of $T_{(R, t)} \mathrm{SE}_{3}$, we compute

$$
\mathcal{R}_{(R, t)}\left(0_{(R, t)}\right)=\left(\begin{array}{cc}
R & t \\
0 & 1
\end{array}\right)\left(\begin{array}{cc}
e^{R^{-1} 0_{R}} & R^{-1} 0 \\
0 & 1
\end{array}\right)=\left(\begin{array}{cc}
R & t \\
0 & 1
\end{array}\right) I_{4}=(R, t) .
$$

With the computation of the differential $D \mathcal{R}_{R}^{\mathrm{SO}_{3}}\left(0_{R}\right)[\zeta]$ in Example 2.3.4 we conclude for some tangent vector $(\zeta, u)=\left(R \hat{\xi}_{\zeta}, u\right) \in T_{(R, t)} \mathrm{SE}_{3}$

$$
\begin{aligned}
D \mathcal{R}_{(R, t)}\left(0_{(R, t)}\right)[(\zeta, u)] & =\left.\lim _{h \rightarrow 0} \frac{1}{h}\left(\mathcal{R}_{(R, t)}(\eta+h \zeta, s+h u)-\mathcal{R}_{(R, t)}(\eta, s)\right)\right|_{(\eta, s)=\left(0_{R, 0}\right)} \\
& =\left(\begin{array}{cc}
R & t \\
0 & 1
\end{array}\right)\left(\begin{array}{cc}
\hat{\xi}_{\zeta} & \lim _{h \rightarrow 0} \frac{1}{h} R^{-1} h u \\
0 & \lim _{h \rightarrow 0} \frac{1}{h}
\end{array}\right) \\
& =\left(\begin{array}{cc}
R & t \\
0 & 1
\end{array}\right)\left(\begin{array}{cc}
\hat{\xi}_{\zeta} & R^{-1} u \\
0 & 0
\end{array}\right) \\
& =\left(\begin{array}{cc}
R \hat{\xi}_{\zeta} & R R^{-1} u+0 t \\
0 & 0
\end{array}\right) \\
& =(\zeta, u) .
\end{aligned}
$$

This proves that the function defined in (2.20) is indeed a retraction for $\mathrm{SE}_{3}$ in the sense of Definition 2.3.1. We can use this retraction to define local updates in $\mathrm{SE}_{3}$.

Definition 2.3.8. For $T_{0}=\left(R_{0}, t_{0}\right) \in \mathrm{SE}_{3}$ and local coordinates

$$
\xi=\binom{\omega}{v} \in \mathbb{R}^{6}
$$

with $\omega=\alpha a \in \mathbb{R}^{3}$ and $v \in \mathbb{R}^{3}$ as described above, we define the local update $\oplus$ of $T_{0}$ by the local coordinates $\xi$ as

$$
T_{0} \oplus \xi:=\mathcal{R}_{T_{0}}\left(R_{0} \hat{\omega}, v\right)
$$

Now we can proceed as in Section 2.3.3. Let $h: \mathrm{SE}_{3} \rightarrow \mathbb{R}^{n}$ be an estimation function. To find

$$
\begin{equation*}
T^{*}=\underset{T \in \mathrm{SE}_{3}}{\arg \min }\|h(T)-z\|^{2} \tag{2.21}
\end{equation*}
$$

for some measurement $z$, we compute the step

$$
\delta^{(t)}=\underset{\xi \in \mathbb{R}^{6}}{\arg \min }\left\|h\left(T^{(t)} \oplus \xi\right)-z\right\|^{2}
$$

to update the current estimate $T^{(t)}$ as $T^{(t+1)}=T^{(t)} \oplus \delta^{(t)}$.

### 2.3.5. Optimizing a Factor Graph

Now that we know how to optimize for poses in $\mathrm{SE}_{3}$, we examine how to optimize a factor graph. In the pose estimation problem described in Section 3.2, we will combine these optimization approaches by optimizing for poses of different objects in $\mathrm{SE}_{3}$ that are linked by a factor graph, simultaneously. This brief introduction to factor graph optimization is based on [13, Sections 1.6, 1.7, and 2.2].

Let $\mathcal{F}=(V, F, E)$ be a factor graph with variables $V$, factors $F$ and edges $E \subseteq V \times F$. Let furthermore $f(X)=\prod_{j} f_{j}\left(X_{j}\right)$ be the global function of the factor graph, defined by the product of the factors $f_{j} \in F$. In general, we can pose the question, which variable assignment $X^{\text {max }}$ maximizes the global function $f$, i.e.

$$
X^{\max }=\underset{X}{\arg \max } f(X)=\underset{X}{\arg \max } \prod_{j} f_{j}\left(X_{j}\right)
$$

Depending on the structure of $\mathcal{F}$ and the factors $f_{j}$, we can make certain reductions to the problem.

Let the factors of $\mathcal{F}$ denote probability densities as in Example 2.2.7. Thus, the global function $f(X)$ of this factor graph is some joint probability density $p(X, Z)$ for unknown states $X$ and given observations and assumptions $Z$. Lemma 2.2 .5 states that maximizing the joint probability density $p(X, Z)$ gives the same argument $X$ as maximizing the posterior density $p(X \mid Z)$. The maximum a posteriori estimate $X^{\text {MAP }}$, in turn, is what we are looking for if we search for the state $X$ that is most likely under certain preconditions, assumptions, and observations $Z$.

Let's assume that the factors $f_{j}\left(X_{j}\right)$ are probability densities of some multivariate Gaussian distribution. As a reminder, the multivariate Gaussian distribution $\mathcal{N}(\mu, \Sigma)$ is given by the density

$$
f(x)=\frac{1}{\sqrt{(2 \pi)^{n} \operatorname{det}(\Sigma)}} e^{-\frac{1}{2}(x-\mu)^{T} \Sigma^{-1}(x-\mu)}
$$

where $x, \mu \in \mathbb{R}^{n}$ and the covariance matrix $\Sigma \in \mathbb{R}^{n \times n}$ [30, Chapter I, Section 5]. In particular, the factors are proportional

$$
\begin{equation*}
f_{j}\left(X_{j}\right) \propto e^{\left.-\frac{1}{2}\left(h\left(X_{j}\right)-z_{j}\right)^{T} \Sigma_{j}^{-1}\left(h\left(X_{j}\right)-z_{j}\right)\right)}=e^{\left.-\frac{1}{2} \| h\left(X_{j}\right)-z_{j}\right) \|_{\Sigma_{j}}^{2}} \tag{2.22}
\end{equation*}
$$

with the notation $\left.\left(h\left(X_{j}\right)-z_{j}\right)^{T} \Sigma_{j}^{-1}\left(h\left(X_{j}\right)-z_{j}\right)=\| h\left(X_{j}\right)-z_{j}\right) \|_{\Sigma_{j}}^{2}$ from [13, Section 2.2]. The measurements $z_{j}$ and the estimation function $h$ are used as in Section 2.3.3. Thus, the error $h\left(X_{j}\right)-z_{j}$ of the estimation function $h$ regarding the measurement $z_{j}$ is normally distributed around the mean 0 with the covariance matrix $\Sigma_{j}$.

With the considerations above, we can compute

$$
\begin{aligned}
& X^{\mathrm{MAP}} \stackrel{2.2 .5}{=} \underset{X}{\arg \max } p(X, Z) \\
&=\underset{X}{\arg \max } \prod_{j} f_{j}\left(X_{j}\right) \\
& \stackrel{(1)}{=} \underset{X}{\arg \max } \prod_{j} e^{-\frac{1}{2}\left\|h_{j}\left(X_{j}\right)-z_{j}\right\|_{\Sigma_{j}}^{2}} \\
& \stackrel{(2)}{=} \underset{X}{\arg \max } \log \left(\prod_{j} e^{-\frac{1}{2}\left\|h_{j}\left(X_{j}\right)-z_{j}\right\|_{\Sigma_{j}}^{2}}\right) \\
&=\underset{X}{\arg \max }-\frac{1}{2} \sum_{j}\left\|h_{j}\left(X_{j}\right)-z_{j}\right\|_{\Sigma_{j}}^{2} \\
& \stackrel{(3)}{=} \underset{X}{\arg \min } \sum_{j}\left\|h_{j}\left(X_{j}\right)-z_{j}\right\|_{\Sigma_{j}}^{2}
\end{aligned}
$$

The equality in (1) holds since the maximal argument does not change if the objective function is multiplied by a constant positive factor. For the equality in (2), we used that the natural logarithm is a strictly increasing function and, thus, preserves the maximal argument. To obtain the equality in (3), we used that the $\arg \max$ of some function $g$ multiplied with a negative constant factor changes to the $\arg \min$ of $g$.

For $X_{j}$ in some manifold, the resulting optimization problem

$$
\begin{equation*}
X^{\mathrm{MAP}}=\underset{X}{\arg \min } \sum_{j}\left\|h_{j}\left(X_{j}\right)-z_{j}\right\|_{\Sigma_{j}}^{2} \tag{2.23}
\end{equation*}
$$

can be solved with the methods discussed in Sections 2.3.1 and 2.3.4 similar to the optimization problem (2.21).
Remark 2.3.9. The later used library GTSAM [11] uses this optimization scheme for optimization on manifolds with factor graphs as documented in [12].

### 2.4. Geometric Algorithms and Data Structures

The loading edge detection problem, described in Section 3.1, raised various geometric issues. The data generated by the sensors was given as point clouds. Section 2.4.1 discusses point cloud manipulation techniques and a data structure that allows to efficiently store, process, and find points in the point cloud.

In Section 2.4.2, we describe and analyze the classic geometric algorithm RANSAC in detail. This algorithm is used several times in different variants in the implementation of the loading edge detection problem.

### 2.4.1. Point Cloud Processing

The most common ways to represent a geometric object in computer vision, computer graphics, or computer-aided geometric design are polygon meshes and point clouds [41].

For more information on polygon meshes and data structures to represent them, see for instance [57]. In this project, we work with point clouds in the loading edge detection problem and mainly with poses of objects represented by one element of $\mathrm{SE}_{3}$ in the pose estimation problem.
Here, we focus on point clouds and their processing and manipulation. According to [41], a point cloud is a set of points (in our case in $\mathbb{R}^{3}$ ) that represents the surface of one or more objects. They can be generated using for example LiDAR sensors or ZED Cameras (see Chapter 3). Sometimes, models of objects generate these data because they are easier to obtain than real data.

In the preprocessing of point clouds, some kind of downsampling is used to reduce the number of points and, consequently, increase the speed of computations on this point cloud. There are several types of downsampling. Some libraries like the open-source Python and C++ package Open3D provide a variety of point cloud manipulating functions, including downsampling functions as described in [64].


Figure 2.6.: Point cloud of a truck downsampled with voxel downsampling for different voxel sizes $x$.

The first downsampling method discussed here is voxel downsampling. Voxel is short for volumetric pixel and can be seen as a pixel in three dimensions [8]. When a point cloud is downsampled by voxel downsampling, the considered space is subdivided into a grid of voxels of some fixed size $x$ (i.e. $x$ gives the side length of the cubes representing the voxels), and the points of the point cloud are assigned to the voxel they are located in. A voxel $V$ containing the points $p_{1}, p_{2}, \ldots, p_{m}$ then generates one point in the new point cloud by
averaging its assigned points as

$$
\frac{1}{m} \sum_{i=1}^{m} p_{i}
$$

The resulting point cloud consists of more uniformly distributed looking points that mitigate real data noise to some extent. Regions with a high point density are thinned out compared to regions with a low point density. This new point cloud is highly dependent on the parameter $x$ of the size of a voxel. If $x$ is too small, almost no downsampling happens because there are no voxels with a high number of points in them. If $x$ is too large, the point cloud can lose some of its characteristic features as the extreme example of just one big voxel shows. In Figure 2.6, a point cloud of a truck before and after downsampling is shown.

Another way to downsample a point cloud is to randomly select every $n$-th point from the original point cloud to create the downsampled point cloud. By using this method, dense regions of a point cloud remain dense compared to sparse regions. It depends on the actual use case whether this method is preferred over voxel downsampling.

Real data usually comes with unwanted noise. Sometimes the depth of a point computed from camera images is too far away from the real depth due to some errors. Especially points in the background of a scene are often poorly matched. To remove these outliers, one can search the neighborhood of each point. If the number of points in the ball with radius $r$ around a point $p$ is below a certain threshold $N$, the point is considered an outlier and is removed from the point cloud. The parameters $r$ and $N$ have to be chosen according to the point cloud, depending on how the point cloud was generated or already downsampled.

If we know which region of a point cloud contains the desired information, we can cut the point cloud accordingly, for instance, remove all points that lie below a plane or outside of a ball with a certain radius and center.

In practical applications, a combination of these methods is chosen to obtain a point cloud in the desired format, ensuring that algorithms deliver correct results and work efficiently. In Section 3.1.1, we explain the downsampling techniques employed in the loading edge detection algorithm.

Some downsampling techniques as well as our implementation of the loading edge detector need an efficient data structure for organizing point clouds. For example, if we want to find all points in a point cloud $\mathcal{C}$ that lie within a certain distance from a reference point $p$ naively, we just iterate over all points in $\mathcal{C}$, compute the distance, and check if this distance is below a certain threshold. Computing the neighbors of all $n$ points in $\mathcal{C}$ takes in total $\Theta\left(n^{2}\right)$ time, which is pretty bad. Thus, we use and briefly describe $k d$-trees, introduced by Bentley in [4].

This is a data structure for $k$-dimensional data. Here, we deal with 3-dimensional points $p=\left(p_{1}, p_{2}, p_{3}\right)^{T} \in \mathbb{R}^{3}$, but this concept generalizes to $k$ dimensions easily. First, we split the set of points according to their first coordinate: We take a plane parallel to the second and third axis such that there is approximately an equal number of points on both sides of the plane. Then, we divide the set of points $\mathcal{C}$ into two sets $\mathcal{C}_{1}, \mathcal{C}_{2}$ for the two sides of the plane. The set $\mathcal{C}_{1}$ is forwarded to the left child of the root and the set $\mathcal{C}_{2}$ to the right child. We split $\mathcal{C}_{1}$ according to the second coordinate of the points (we divide the space with a plane parallel to the first and third axis) into the sets $\mathcal{C}_{11}$ and $\mathcal{C}_{12}$. Then,
we split $\mathcal{C}_{11}$ according to their third coordinate (we divide the space with a plane parallel to the first and second axis) into the sets $\mathcal{C}_{111}$ and $\mathcal{C}_{112}$. After that, we start again by dividing the points according to their first coordinate. We continue for each set $\mathcal{C}_{n_{1} n_{2} \ldots n_{l}}$ with this procedure until each point can be identified uniquely. This can be seen as a higher dimensional generalization of binary search trees. Querying a $k d$-tree with an axis-aligned search region takes $O\left(n^{1-\frac{1}{k}}+m\right)$ time, where $n$ is the number of points, $k$ the dimension of the data, and $m$ the number of points in the output of the range query.

A $k d$-tree can be used to estimate the normal vectors of points in a point cloud. Assuming that the points in the point cloud represent the surface of an object, it makes sense to equip the points with the normal vector of the represented surface at the respective positions. The Open3D function estimate_normals realizing these considerations, uses a $k d$-tree to find all points that are close to the considered point $p$ [64]. Let $N_{p}$ be the set of neighbors of $p$ found with the $k d$-tree. To compute the normal of that point, we first compute the covariance matrix as

$$
S=\frac{1}{n} \sum_{x \in N_{p}}(x-\mu)(x-\mu)^{T}
$$

where $n$ denotes the number of points and $\mu=\frac{1}{n} \sum_{x \in N_{p}} x$ the center of the points in $N_{p}$. The two eigenvectors $v_{1}$ and $v_{2}$ to the two largest eigenvalues of $S$ define the two principal components of $N_{p}$ [15, Section 10.13.1]. Thus, the normal vector $n_{p}$ of the point $p$ is estimated as the cross-product $v_{1} \times v_{2}$ of the two principal directions.

### 2.4.2. RANSAC

The fitting of lines, planes, circles, parabolas, or many other simple geometric objects is an important issue, that appears in many real-life applications. It is a geometric optimization problem: Fitting a parabola given by $f(x)=a x^{2}+b x+c$ in a set of data points $\left\{\left(x_{i}, y_{i}\right)_{i=1}^{N}\right\} \subseteq \mathbb{R}^{2}$ using least-squares regression is the task of minimizing the sum of squared errors

$$
\sum_{i=1}^{N}\left(y_{i}-f\left(x_{i}\right)\right)^{2}
$$

with respect to the parameters $a, b, c \in \mathbb{R}[49]$.
Fitting a plane given of the equation $E: a x+b y+c z+d=0$ in a set of data points $\left\{\left(x_{i}, y_{i}, z_{i}\right)_{i=1}^{N}\right\} \subseteq \mathbb{R}^{3}$ using orthogonal regression is the task of minimizing the sum of squared orthogonal distances

$$
\sum_{i=1}^{N} d\left(E, p_{i}\right)^{2}
$$

with respect to the parameters $a, b, c, d \in \mathbb{R}$, where $d\left(E, p_{i}\right)$ denotes the orthogonal distance of the point $p_{i}=\left(x_{i}, y_{i}, z_{i}\right)$ to the plane $E$ [29, Section 4.7.1].

These approaches as well as many other geometric estimation approaches consider outlier points to some degree. In Section 3.1 we want to fit lines in a 3D point cloud of a scene captured with sensors and processed with some functions. The line should estimate the loading edge of a truck. In this scenario, the two approaches mentioned above could not deliver a reasonable result since they try to minimize the error to points that have nothing
to do with the loading edge. Here we use the RANSAC algorithm proposed by Fischler and Bolles in [19].

RANSAC stands for random sample consensus and is a model-fitting method that is robust with respect to outliers. Contrary to the regression techniques described above, outliers that do not resemble the optimal solution are eliminated instead of trying to adapt the solution to them. We start with a geometric model that can be defined by at least $m$ points. Then, $m$ points are randomly sampled from the data set. For the instance of the model defined by these points, the total number of data points that lie close enough to the model is counted. The algorithm repeats this process and keeps the best solution with respect to the number of votes. Algorithm 1 shows a simple pseudo-code of this approach.

```
Algorithm 1 Random Sample Consensus (RANSAC)
    Input: point cloud \(\mathcal{C}\), model tolerance \(\varepsilon\), maximum number of iterations \(N\)
    Output: best model parameters \(M\)
    \(M_{\text {best }} \leftarrow\) None
    \(n_{\text {best }} \leftarrow 0\)
    for \(i \leftarrow 1\) to \(N\) do
        randomly sample \(m\) points from \(\mathcal{C}\)
        \(M \leftarrow\) parameters for the model defined by the sampled points
        \(n \leftarrow\) number of points in \(\mathcal{C}\) with a distance \(<\varepsilon\) to the model with parameters \(M\)
        if \(n>n_{\text {best }}\) then
            \(n_{\text {best }} \leftarrow n\)
            \(M_{\text {best }} \leftarrow M\)
        end if
    end for
    return \(M_{\text {best }}\)
```

Examples of simple use-cases of RANSAC are line fitting, where a line is defined by two points, plane fitting, where a plane is defined by three non-collinear points, and circle fitting, where a circle is defined by three non-collinear points. Some models underlie restrictions regarding the defining points. There is the case of, for instance, a plane that is not sufficiently defined by three collinear points. This can be fixed by sampling more points until there are three non-collinear points given. A circle on the other hand cannot be defined by adding more sample points if the initial three points are collinear (if no circle of infinite radius, i.e. a line, is allowed). This situation could be handled by keeping two points and resampling the third one until they are not collinear. These model-specific issues can be addressed when the parameters for the model are computed.

We will choose the model tolerance $\varepsilon$ as seen in Algorithm 1 according to the scene we are working on. In particular, for the loading edge detection, $\varepsilon$ is chosen with respect to the accuracy of the generated point cloud. If the loading edge is nearly a straight line, $\varepsilon$ can be very small. Otherwise, $\varepsilon$ has to be chosen bigger, such that all points on the loading edge are considered for a good RANSAC approximation. In Section 3.1 we chose a tolerance of 0.08 meters.

The maximum number of iterations $N$ is important for the running time of the algorithm.

For finding a line in a point cloud of 1000 points, there are already $\binom{1000}{2}=499500$ possible point pairs. This number grows like $\Theta\left(N^{2}\right)$. So, $N$ has to be chosen small enough for the algorithm to run efficiently but big enough that a good approximation can be found.

Fischler and Bolles, who first described the RANSAC algorithm, propose in [19, Section II.B.] a way of estimating the maximum number of iterations $N$ required to get a good solution with a certain probability. We adapt this approach here. Let us assume that we have a set $\mathcal{C}$ of $c$ points and the true optimal solution $S \subseteq \mathcal{C}$ contains $n \leq c$ points. Furthermore, let $m \leq n$ be the number of model parameters of the model we try to fit into the set $\mathcal{C}$. We will call the $n$ points in $S$ inlier points or just inliers.

Let $K_{m}$ be the random variable giving the number of iterations of the for-loop of RANSAC until we find a set of model parameters $M$ defining the solution $S$. Let us assume that this is achieved if and only if the $m$ chosen points are in the solution set, i.e. $M \subseteq S$. For a small tolerance $\varepsilon$, this gives a good approximation. Then the expected value $\mathbb{E}\left(K_{m}\right)$ of the number of trials is given by

$$
\mathbb{E}\left(K_{m}\right)=\sum_{k \geq 1} \mathbb{P}\left(K_{m}=k\right) k
$$

where $\mathbb{P}\left(K_{m}=k\right)$ denotes the probability that the correct solution is obtained in the $k$-th trial for the first time. Let

$$
\begin{equation*}
q_{m}:=\frac{\binom{n}{m}}{\binom{c}{m}}=\frac{n!}{m!(n-m)!} \frac{m!(c-m)!}{c!}=\frac{n!(c-m)!}{c!(n-m)!} \tag{2.24}
\end{equation*}
$$

be the probability that $m$ randomly chosen points in $\mathcal{C}$ lie in the set of $S$. Then the probability $\mathbb{P}\left(K_{m}=k\right)$ is obtained by $k-1$ unsuccessful attempts followed by a successful trial as

$$
\mathbb{P}\left(K_{m}=k\right)=\left(1-q_{m}\right)^{k-1} q_{m}
$$

Now we can compute the expected value as

$$
\begin{aligned}
\mathbb{E}\left(K_{m}\right) & =\sum_{k \geq 1}\left(1-q_{m}\right)^{k-1} q_{m} k \\
& \stackrel{(1)}{=} q_{m} \sum_{k \geq 1} k \bar{q}_{m}^{k-1} \\
& \stackrel{(2)}{=} q_{m}\left(\sum_{k \geq 0} \bar{q}_{m}^{k}\right)^{\prime} \\
& \stackrel{(3)}{=} q_{m}\left(\frac{1}{1-\bar{q}_{m}}\right)^{\prime} \\
& =q_{m}(-1) \frac{-1}{\left(1-\bar{q}_{m}\right)^{2}}=\frac{q_{m}}{q_{m}^{2}}=\frac{1}{q_{m}}
\end{aligned}
$$

In the above computation, the equality in (1) is obtained by defining $\bar{q}_{m}:=1-q_{m}$. The equalities in (2) and (3) follow from the rules of differentiating formal power series and the formal power/Laurent series identity $\sum_{i \geq 0} X^{i}=\frac{1}{1-X}$ [51, Section 8.4].

The variance $\mathbb{V}\left(K_{m}\right)$ of the number of iterations till success is then given by

$$
\begin{aligned}
\mathbb{V}\left(K_{m}\right) & =\mathbb{E}\left(K_{m}^{2}\right)-\mathbb{E}\left(K_{m}\right)^{2} \\
& =\sum_{k \geq 1} \mathbb{P}\left(K_{m}=k\right) k^{2}-\left(\sum_{k \geq 1} \mathbb{P}\left(K_{m}=k\right) k\right)^{2} \\
& =\sum_{k \geq 1}\left(1-q_{m}\right)^{k-1} q_{m} k^{2}-\left(\frac{1}{q_{m}}\right)^{2} \\
& =q_{m} \sum_{k \geq 1} k(k-1+1) \bar{q}_{m}^{k-1}-\frac{1}{q_{m}^{2}} \\
& \stackrel{(4)}{=} q_{m} \bar{q}_{m} \sum_{k \geq 2} k(k-1) \bar{q}_{m}^{k-2}+q_{m} \sum_{k \geq 1} k \bar{q}_{m}^{k-1}-\frac{1}{q_{m}^{2}} \\
& \stackrel{(5)}{=} q_{m} \bar{q}_{m}\left(\sum_{k \geq 0} \bar{q}_{m}^{k}\right)^{\prime \prime}+q_{m}\left(\sum_{k \geq 0} \bar{q}_{m}^{k}\right)^{\prime}-\frac{1}{q_{m}^{2}} \\
& \stackrel{(6)}{=} q_{m} \bar{q}_{m}(-1) \frac{-2}{\left(1-\bar{q}_{m}\right)^{3}}+\frac{1}{q_{m}}-\frac{1}{q_{m}^{2}} \\
& =\frac{2-2 q_{m}}{q_{m}^{2}}-\frac{1-q_{m}}{q_{m}^{2}}=\frac{1-q_{m}}{q_{m}^{2}} .
\end{aligned}
$$

In (4), the index can be shifted to $k \geq 2$ since for $k=1$ the summand $k(k-1) \bar{q}_{m}^{k-2}$ is 0 . The equalities in (5) and (6) work as in (2) and (3) above.

This gives just a rough idea of the magnitude of a reasonable number of iterations $N$. We can also choose $N$ such that all chosen points lie in the solution set $S$ in at least one of $N$ iterations with a certain probability $p$ as in [29, Section 4.7.1]. This is the same as failing to choose all points in $S$ in all $N$ iterations with a probability of $1-p$. Thus, with the notation above, the equation

$$
1-p=\left(1-q_{m}\right)^{N}
$$

defines $N$ for given $p$ and $q_{m}$. Taking the $\log$ of this equation and dividing by $\log \left(1-q_{m}\right)$ gives the equivalent equation (under the assumption $p, q_{m} \notin\{0,1\}$ )

$$
\begin{equation*}
N=\frac{\log (1-p)}{\log \left(1-q_{m}\right)} . \tag{2.25}
\end{equation*}
$$

Lastly, we can simplify the definition of $q_{m}$ as seen in (2.24). Usually, we do not know the exact numbers $c$ of total points and $n$ of inlier points. Instead, we can estimate the percentage of inlier points compared to the total number of points in $\mathcal{C}$. Furthermore, $m$ is typically small (e.g. 2 for lines, 3 for planes) compared to the total number of points and inliers. Thus, we can assume that the probability of choosing an inlier remains constant for all chosen points. Let $r$ be this probability (approximately $\frac{n}{c}$ with the notation above), then $q_{m}$ simplifies to

$$
q_{m}=r^{m} .
$$

Example 2.4.1. Let $\mathcal{C} \subseteq \mathbb{R}^{2}$ be a point cloud as in Figure 2.7. We do not want to find a regression line but rather the longest line of points in the point cloud. Thus, RANSAC is an appropriate approach. About $\frac{1}{3}$ of the points lie on the longest line. The expected number of trials to success is then given by

$$
\mathbb{E}\left(K_{2}\right)=\frac{1}{\left(\frac{1}{3}\right)^{2}}=9
$$

and the variance and standard deviation by

$$
\mathbb{V}\left(K_{2}\right)=\frac{1-\frac{1}{9}}{\left(\frac{1}{9}\right)^{2}}=72 \quad \text { and } \quad \sqrt{\mathbb{V}\left(K_{2}\right)}=6 \sqrt{2} \approx 8.49 .
$$

To get approximately this line with a probability of at least $99 \%$, we perform RANSAC with a maximum number of $N=40$ iterations according to

$$
\frac{\log (1-0.99)}{\log \left(1-\frac{1}{9}\right)}=\frac{\log (0.01)}{\log \left(\frac{8}{9}\right)} \approx 39.10
$$



Figure 2.7.: RANSAC line detection example.

# 3. Composite Object Detection in a Loading Scenario of a Truck 

This thesis is part of a bigger project at the Austrian Institute of Technology (AIT), dealing with the autonomous loading of trucks. Here, we address two specific problems of a loading scenario.

The first issue is to detect the loading edge of a truck in a point cloud that is generated by LiDAR sensors or depth cameras. Our solution method includes several geometric considerations on these points and is presented and analyzed in Section 3.1. The essential parts of the code of our implementation can be found in Appendix A.

In the second and central part of this chapter, we view the whole truck as an object that is composed of simpler objects, namely wheels, lights, and the loading platform (respectively the two loading edges on both sides). In a loading scenario, a forklift equipped with sensors should be able to estimate its own pose relative to the truck, given only a rough approximation of the configuration of the truck, i.e. the relative poses of the different parts of the truck to one another. This pose estimation approach is examined in Section 3.2. In Appendix B, we present the crucial parts of our implementation for solving this pose estimation problem.

The algorithms and methods used in this chapter are based on the mathematical foundation examined in Chapter 2. In Section 3.1, we use point cloud manipulation techniques and the RANSAC algorithm of Section 2.4. The pose estimation problem of Section 3.2 is based on factor graphs (see Section 2.2) and optimization on manifolds (see Section 2.3).

All data used in this work come from real recordings of a truck. We used a MAN TGS 26.440 truck [44], a PALFINGER BM 214 truck-mounted forklift (also called Crayler) [50], and a ZED $2 i$ Camera [54], all provided by Palfinger and the AIT. Figure 3.1 displays the used equipment. The truck was used in both the loading edge detection of Section 3.1 and the pose estimation problem of Section 3.2 as the composite object to be detected. The ZED Camera was mounted on the top of the forklift and the forklift drove around the truck, observing it with the camera. Above the ZED Camera, we installed a LiDAR sensor (short for Light Detection and Ranging), which generates a point cloud of its surroundings with laser scanning. The data obtained from the ZED Camera and the LiDAR sensor could be combined to get better data. Here, we only used the ZED Camera for recording.

The ZED Camera is equipped with two cameras, recording at the same time. Thus, it captures stereo images enabling depth perception. The depth data is computed by bundle adjustment. This technique compares matching points in the left and right image to compute the corresponding 3D point by optimizing a nonlinear least squares problem [7]. The internal coordinate system of the ZED Camera depends on the application and the software used to generate the data. In Remark 3.1.1, we describe the two variants used in this work. We refer to [53] for further information on the coordinate frames of this camera.


Figure 3.1.: Truck, autonomous forklift, and ZED Camera used in our experiments.

### 3.1. Loading Edge Detection

When a truck is being loaded from the side, the essential part of the truck to detect is the loading edge, i.e. the side boundary of the loading platform. In Figure 3.2 one can see a point cloud of a truck recorded with a ZED Camera, where the loading edge is detected with our proposed algorithm and marked in red.

The presented method of loading edge detection relies on the geometric properties of points on the loading edge. Additionally, it uses initial estimates of the height of the loading edge above the ground as well as the height and angle of the camera. These parameters are usually easy to get before using the algorithm with some knowledge about the setting of the camera attached to the autonomous forklift and the truck whose loading edge has to be detected. The more accurate these estimates are, the faster the algorithm gets, because the part of the point cloud where the loading edge could be located can be restricted accordingly.

In light of the pose estimation task described in Section 3.2, we are not only interested in the pose of the loading edge for the autonomous forklift to know where to place the load. The beginning of the left loading edge will mark the origin of the world coordinate frame in which the sensor should locate itself. The direction of the $x$-axis of the world frame will be defined by the left loading edge and the $z$-axis will point upwards, orthogonal to the loading platform. The $y$-axis will be chosen such that the world frame forms a right-handed
coordinate system. In Figure 3.12 the coordinate system of the world frame is marked with three arrows defining the axes.

Thus, this algorithm also tries to find the edge between the loading platform and the rear wall to get the beginning of the loading edge and to define the plane incident to the loading platform and therefore the direction of the $z$-axis. In Appendix A the important parts of the code of our approach to the loading edge detection problem can be found.

In Section 3.1.1 we describe how we solved and implemented the loading edge detection problem and briefly discuss different approaches for edge detection algorithms. Then, we analyze our code and the parameters of our code in Section 3.1.2.


Figure 3.2.: Point cloud of a truck with the detected loading edge marked in red.

### 3.1.1. Description of the Algorithm

The presented algorithm deals a lot with point cloud processing and manipulation. The Python and C++ library Open3D [64] is specialized in 3D Data Processing and was used in the version 0.17.0. It is compatible with the Python package NumPy [28] which speeds up computation and makes arrays of any shape easy to handle. In addition to these two packages, the python time package was used for measuring the performance of the algorithm in terms of running time and comparing different parameter combinations.

We chose an object-oriented approach. For each point cloud, we want to find the loading edge in, an instance of the class LoadingEdgeDetection is created and initialized with this Open3D point cloud, an estimated height and angle of the sensor(s) (ZED Camera and/or LiDAR sensor), and an estimated height of the loading platform. Furthermore, some other parameters can be set optionally when initializing a class instance, otherwise, their default values are used. They will be described below.

When an instance of the LoadingEdgeDetection class is initialized, the up-vector $u^{*}$ ( $z$-axis of the world frame) of the scene is estimated in the sensor coordinate frame from the given depth angle of the camera. This is equivalent to finding a rotation in $\mathrm{SO}_{3}$ that
transforms the $z$-axis of the sensor frame to the $z$-axis of the world frame. Here, the sensor frame is defined as a right-handed coordinate system with the $x$-axis pointing in the direction the sensor is looking at, the $y$-axis pointing to the left, and the $z$-axis pointing upwards. Since $u^{*}$ is only a rough estimation of the real up-vector of the scene, we assume that the camera is just tilted to the front. Therefore, we only need a rotation around the $y$-axis to rotate the $z$-axis of the sensor coordinate frame upwards. Thus, for a given depth angle $\alpha$ of the camera, the up-vector is given by

$$
u^{*}=\left(\begin{array}{ccc}
\cos (-\alpha) & 0 & -\sin (-\alpha) \\
0 & 1 & 0 \\
\sin (-\alpha) & 0 & \cos (-\alpha)
\end{array}\right)\left(\begin{array}{l}
0 \\
0 \\
1
\end{array}\right)=\left(\begin{array}{c}
-\sin (-\alpha) \\
0 \\
\cos (-\alpha)
\end{array}\right)
$$

Remark 3.1.1. The choice of the sensor frame is dependent on the hardware and software used to obtain the data. There is no general convention for sensor coordinate frames. Here, we obtained the data with a ZED Camera using the camera frame standard of the robot operating system (ROS). In Section 3.2, we use a sensor frame where the sensor looks in negative $z$-direction, the $y$-axis points upwards, and the $x$-axis to the right since the data used in that algorithm are obtained by Blender which is based on the OpenGL standard camera definition, where this is the standard sensor frame [53].

Working with real data presents some difficulties. First, we have only limited control over the amount of data generated by the sensors. Here, with over 2.7 million points generated by the ZED Camera, computation would take way too long for practical purposes. Thus, we need to preprocess the point cloud with techniques described in Section 2.4.1. Figure 3.3 shows the downsampling process used here.

We start by performing voxel downsampling using a voxel size of 0.015 meters reducing the point cloud to roughly 700,000 points. In the next step, outliers are removed, such that around 600,000 points remain. For a point cloud generated by a ZED Camera, a point $p$ is considered an outlier if there are less than 30 points within a radius of 0.05 meters around $p$. Lastly, we use the estimated height of the loading platform and the estimated direction of $u^{*}$ to get an approximate pose of the plane $L$ that is incident to the loading platform. Then we cut the point cloud around this plane and remove every point that is too far away from $L$. About 200,000 points remain. The parameter search_width defines the width of the remaining strip of points. Besides lowering the number of points and thus lowering the computation time, the last step has another effect: We cut away a lot of points of the scene that could be detected as edge points by the algorithm, for instance, the edge between the floor and a wall is an edge that could be declared as loading edge falsely. Thus, search_width should especially be chosen small enough to cut the floor away. We refer to the point cloud obtained after these downsampling steps as $\mathcal{C}$.

Now the preprocessing of the point cloud is finished and the search for the loading edge can start. As mentioned above, the goal is to find the beginning of the loading edge, i.e. the loading edge $e_{l}$ and the edge $e_{w}$ between the loading platform and the rear wall of the truck, if possible. In Figure 3.2 these two edges are marked in green and red. The algorithm does not work with semantic information about $e_{l}$, in particular, only the geometric properties of the points are used to find the loading edge. To be precise, the algorithm first looks for the longest edge $e_{1}$ in the point cloud $\mathcal{C}$ that is approximately orthogonal to the estimated
up-vector $u^{*}$ with the function find_longest_edge(). In the second step, the function find_orthogonal_line() searches for an edge $e_{2}$ that is approximately orthogonal to $e_{1}$ and $u^{*}$. Depending on the camera's pose and the resulting perspective, either $e_{l}$ or $e_{w}$ is found first.

Let us assume that every edge detection and every estimation of angles and heights works perfectly (or sufficiently well). Then the algorithm delivers the correct result or approximation if the longest visible edge $e_{1}$ in the point cloud is in $\left\{e_{l}, e_{w}\right\}$ and the longest visible edge orthogonal to $e_{1}$ and $u^{*}$ is also in $\left\{e_{l}, e_{w}\right\}$.

To find the edge $e_{1}$, we first compute possible edge points in $\mathcal{C}$. The method that works best among all methods considered here in this context uses the estimated normals of the points in the point cloud. The normals of points in a point cloud can be estimated as described in Section 2.4.1. Figure 3.4 shows the edge points detected by this method. Alternative approaches to finding edge points are discussed at the end of this section.

To get edge points, we compute the $k d$-tree of the points in $\mathcal{C}$ (see Section 2.4.1 for more about $k d$-trees) and estimate the normals of the points. To further accelerate the computation, the point cloud $\mathcal{C}$ is then downsampled one more time. Now, we use the uniform downsampling method of Open3D with a downsampling factor of 40, which randomly selects every 40 -th point, to get the point cloud $\mathcal{C}^{\prime}$. The roughly 200,000 points in $\mathcal{C}$ reduce to about 5,000 points in $\mathcal{C}^{\prime}$. Now we search for edge points in the point cloud $\mathcal{C}^{\prime}$.


Figure 3.3.: Downsampling process of a point cloud of a truck: First, we apply voxel downsampling, then we remove outlier points, and then we restrict the point cloud to a small area around the loading platform.


Figure 3.4.: The blue points are the edge points that are detected by using the estimated point normals. For illustration purposes, the edge point detection was carried out on $\mathcal{C}$ instead of $\mathcal{C}^{\prime}$.

For each point $p$ in $\mathcal{C}^{\prime}$, let $N_{p} \subseteq \mathcal{C}$ be the set of neighbors of $p$, i.e. all points in $\mathcal{C}$ that lie within a radius of 0.1 meters around $p$. This set can be efficiently obtained by using the $k d$-tree computed for $\mathcal{C}$. Let $N_{p}^{\text {up }} \subseteq N_{p}$ be the set of all points whose normals are (approximately) parallel to the up-vector $u^{*}$ and let $N_{p}^{\mathrm{rest}}:=N_{p} \backslash N_{p}^{\mathrm{up}}$ be the set of all other points. In this approach, we call a point $p$ an edge point as declared in the following definition.

Definition 3.1.2 (Edge Points using Estimated Normals). Concerning the conditions and definitions stated above, we call a point $p$ an edge point if all of the following properties hold.
(i) The normals of the points in $N_{p}^{\mathrm{up}}$ have (approximately) the same direction $v_{p}^{\mathrm{up}}$.
(ii) The normals of the points in $N_{p}^{\text {rest }}$ have (approximately) the same direction $v_{p}^{\text {rest }}$.
(iii) The vectors $v_{p}^{\mathrm{up}}$ and $v_{p}^{\text {rest }}$ are (approximately) orthogonal.
(iv) The sets $N_{p}^{\mathrm{up}}$ and $N_{p}^{\mathrm{rest}}$ are (approximately) the same size.

All these conditions are just approximately because real data as well as the estimated normals are not perfect. This is a reasonable way to define edge points because under these conditions, a point $p$ lies in the intersection of two planes (all points in a plane have the same normal vector). Part (iii) of the definition ensures that the edge found is created by two nearly orthogonal planes, just like the loading edge. This definition of edge points is similar to the edge detection approach described in [60, Section 4.2].

Figure 3.5 visualizes this edge point definition. The arrows indicate the estimated normals for points close to the considered (orange) point $p$. If the neighbors of $p$ lie in a plane, all estimated normals are approximately parallel. If the points in $N_{p}$ lie on a curved
surface, the estimated normals point in various directions. In the last case, $p$ is indeed an edge point. The estimated normals cluster in two approximately orthogonal groups as described above.


Figure 3.5.: Normals for points on planes, curved surfaces, and edges as [60, Figure 3].
We can define the vectors $v_{p}^{\text {up }}$ and $v_{p}^{\text {rest }}$ as

$$
v_{p}^{\mathrm{up}}=\frac{1}{\left|N_{p}^{\mathrm{up}}\right|} \sum_{q \in N_{p}^{\mathrm{up}}} n_{q} \quad \text { and } \quad v_{p}^{\text {rest }}=\frac{1}{\left|N_{p}^{\mathrm{rest}}\right|} \sum_{q \in N_{p}^{\text {rest }}} n_{q},
$$

where $n_{q}$ is the estimated normal vector of a point $q$. The normals $n_{q}$ in $N_{p}^{\mathrm{up}}$ have approximately the same direction if the length of $v_{p}^{\text {up }}$ is close to 1 . So, parts (i) and (ii) are checked by computing $v_{p}^{\mathrm{up}}$ and $v_{p}^{\text {rest }}$ and their lengths. Allowing shorter vectors leads to more accepted edge points. So if (i) and (ii) hold, the points of $N_{p}^{\text {up }}$ and the points of $N_{p}^{\text {rest }}$ lie in a plane each.

Condition (iii) of Definition 3.1.2 is only checked if both vectors $v_{p}^{\mathrm{up}}$ and $v_{p}^{\text {rest }}$ are not the zero-vector. Thus, we can compute the inner product

$$
\begin{equation*}
\left\langle\frac{1}{\left\|v_{p}^{\mathrm{up}}\right\|} v_{p}^{\mathrm{up}}, \frac{1}{\left\|v_{p}^{\mathrm{rest}}\right\|} v_{p}^{\mathrm{rest}}\right\rangle . \tag{3.1}
\end{equation*}
$$

If this inner product is 0 , these vectors are orthogonal. Allowing inner products close to 0 leads to more accepted edge points. This relaxation makes sense since the inner product (3.1) relates to the angle $\alpha$ between the corresponding vectors by the formula [39, Section 6.1 (2)]

$$
\cos \alpha=\frac{\left\langle v_{p}^{\mathrm{up}}, v_{p}^{\mathrm{rest}}\right\rangle}{\left\|v_{p}^{\mathrm{u}}\right\|\left\|v_{p}^{\text {rest }}\right\|}=\left\langle\frac{1}{\left\|v_{p}^{\mathrm{up}}\right\|}\left\|_{p}^{\mathrm{up}}, \frac{1}{\left\|v_{p}^{\text {rest }}\right\|}\right\|_{p}^{\text {rest }}\right\rangle .
$$

If this condition holds, the edge is created by two (nearly) orthogonal planes.
Lastly, part (iv) of Definition 3.1.2 can easily be checked by comparing the number of points in both sets. Allowing small differences in the number of contained points leads to more accepted edge points. This condition ensures that the points accepted as edge points lie sufficiently close to the edge. In an extreme case, the set $N_{p}^{\text {rest }}$ could, for example, contain only one point with all other points belonging to $N_{p}^{\text {up }}$ and thus, under condition (i), lying in a horizontal plane, rather far away from the edge.

The partition into the sets $N_{p}^{\mathrm{up}}$ and $N_{p}^{\text {rest }}$ as well as the checks of the conditions above happen in the functions find_edge_points_normals() and cluster_normals().

Now, we have a set $\mathcal{C}_{\text {edge }} \subseteq \mathcal{C}^{\prime}$ of edge points. The edge $e_{1}$ is defined as the line passing through the most points in $\mathcal{C}_{\text {edge }}$, with a small tolerance allowing points to lie close to the line. This is done by a RANSAC approach as described in Section 2.4.2. In Figure 3.4 about $\frac{1}{3}$ of all edge points lie on the loading edge. Thus, with Formula (2.25), at least 40 iterations of RANSAC are required to find this edge with a probability of $99 \%$.

To find $e_{2}$, the function find_orthogonal_line() searches in $\mathcal{C}_{\text {edge }}$ for the longest edge (approximately) orthogonal to $e_{1}$, using a RANSAC approach. In particular, with RANSAC the line with the most votes in $\mathcal{C}_{\text {edge }}$ is found. If this line is approximately orthogonal to $e_{1}$ and $u^{*}$, it is accepted as $e_{2}$. Otherwise, the points on this line are deleted from $\mathcal{C}_{\text {edge }}$ and the process is started again. The parameter max_lines defines how often this search is restarted. By doing this, we want to avoid accepting an edge orthogonal to $e_{1}$ that consists of too few points, because the edge between the loading platform and the rear wall of the truck does not have to be visible in every analyzed scene. So if no second edge $e_{2}$ is detected within a few iterations, the algorithm just finds one edge.

If a second edge was found, we want to declare a point as the origin of the world frame. Using perfect data, this would be the intersection point of $e_{1}$ and $e_{2}$. Generally, these two lines do not intersect in practice. Let $E_{i}$ for $i=1,2$ be the plane determined by $e_{i}$ and the estimated up-vector $u^{*}$. Let $S_{1}$ be the intersection point $e_{1} \cap E_{2}$ and $S_{2}$ be the intersection point $e_{2} \cap E_{1}$. The origin is then defined as the midpoint $\frac{1}{2}\left(S_{1}+S_{2}\right)$ between the two intersection points.

In the next step, the directions of the $x$-axis and the $y$-axis are defined. The coordinate frame should look as in Figure 3.12, so the loading edge and the edge between the loading platform and the rear wall lie in positive $x$ - respectively $y$-direction of the origin. Again with a RANSAC-like approach, we look for the endpoints of the lines $e_{1}$ and $e_{2}$ in $\mathcal{C}_{\text {edge }}$ : Two edge points are sampled and the pair of points with the greatest distance between them is accepted as the pair of endpoints of the line. This does not have to be the very best solution possible, some good approximation is sufficient. The vector from the origin to the endpoint that is further away defines the direction of the corresponding axis.

As stated above, it is not clear if $e_{l}$ or $e_{w}$ is the line $e_{1}$ that is found first by the algorithm. To differentiate $e_{l}$ from $e_{w}$, we look at a difference between these two edges, that can easily be detected in a point cloud: Directly above the loading edge $e_{l}$ are no points, while directly above $e_{w}$ there are still many points of the rear wall. So both lines are lifted by a few centimeters in the direction of $u^{*}$ and the number of points on the lines before and after the lifting are compared. The line with the smaller quotient of the number of points on the line after and before lifting is declared as the loading edge $e_{l}$.

Let $v_{l}$ and $v_{w}$ be the direction vectors of $e_{l}$ and $e_{w}$. If the truck is seen from the left side, the direction vectors $v_{x}^{\text {left }}, v_{y}^{\text {left }}$, and $v_{z}^{\text {left }}$ are defined as

$$
v_{x}^{\text {left }}:=v_{l}, \quad v_{z}^{\text {left }}:=v_{x}^{\text {left }} \times v_{w}, \quad v_{y}^{\text {left }}:=v_{z}^{\text {left }} \times v_{x}^{\text {left }} .
$$

The order of the factors of the cross-product is important and can be checked with the right-hand rule. If the truck is seen from the right side, this process does not define the origin of the world frame. But concerning the factor graph approach for the pose estimation
problem in Section 3.2, we define the pose of the right loading edge by the coordinate frame where $e_{l}$ defines the $y$-axis instead of the $x$-axis as above. So, let

$$
v_{y}^{\text {right }}:=v_{l}, \quad v_{z}^{\text {right }}:=v_{w} \times v_{y}^{\text {right }}, \quad v_{x}^{\text {right }}:=v_{y}^{\text {right }} \times v_{z}^{\text {right }}
$$

be the direction vectors of the axes if the right loading edge was detected.
If only the loading edge $e_{l}$ was detected, the origin is set to one endpoint randomly. The $z$-axis is defined by the up-vector and the last axis is defined by the cross product of $v_{l}$ and $u^{*}$. Whether the last axis is taken as $v_{l} \times u^{*}$ or as $u^{*} \times v_{l}$ is determined by moving the line of the loading edge a bit to the side to detect on which side of the loading edge the loading platform is located.

This is our approach to solving the loading edge detection problem. A different approach to defining the edge points was attempted in this project: Instead of estimating the normals of all points in the neighborhood $N_{p}$ of a potential edge point $p$, we compute the centroid $c_{p}=\frac{1}{n} \sum_{p^{\prime} \in N_{p}} p^{\prime}$ of the neighbors of $p$ and measure the distance $d\left(p, c_{p}\right)$. This edge detection method is also described in [3]. If this distance is larger than a certain threshold, the point $p$ is declared as an edge point. Figure 3.6 shows the edge points detected in that way. We can see that a lot more points on edge-like structures in the point cloud (especially on things lying on the loading platform) are declared as edge points. Furthermore, points on borders of the point cloud fulfill this property too. Methods like this centroid approach lack the possibility of considering previous knowledge about the edge direction, such as a normal vector to that direction. A possible improvement would be to consider the vector $c_{p}-p$ instead of just the distance $d\left(p, c_{p}\right)=\left\|c_{p}-p\right\|_{2}$. Du summarizes and compares different edge detection techniques in his work [14], including the normal vector approach and the centroid approach.


Figure 3.6.: The blue points are the edge points that are detected by using the centroids of the neighbors.

### 3.1.2. Parametrization and Analysis of the Algorithm

The quality of the result of the loading edge detection algorithm depends on the quality of the data. For a perfectly generated point cloud and exact estimates of the height and angle of the camera, this algorithm delivers the correct result efficiently. LiDAR sensors generate point clouds of good quality. Therefore, a LiDAR sensor would work well together with the loading edge detector. Since the necessary hardware was not yet installed properly, we only have limited access to testing data. We will analyze the influence of different parameters on the algorithm's running time, using a point cloud generated with data from a ZED Camera. The ground truth of the detected loading edge can be seen in Figure 3.7.


Figure 3.7.: Point cloud of the truck, where the right loading edge (green) is detected.

First, we analyze the used voxel size, i.e. the size of the voxel grid, that we use to downsample the point cloud in the preprocessing step. Figure 3.8 displays the running time of the algorithm, the error to the optimal solution, and the success rate for different voxel sizes, where each parameter set was tested 100 times. The lines indicate the mean values of the 100 trials and the shaded regions around the lines indicate the empirical standard deviation. As the voxel size increases, fewer points remain in the point cloud, in which we want to find the loading edge. The running time of operations like finding the neighbors of all points in a $k d$-tree and computing the edge points depends on the number of points in the point cloud. Additionally, the voxel downsampling is more computationally expensive if the voxel size is smaller. Therefore, a larger voxel size results in a shorter running time of the algorithm (as portrayed in Figure 3.8) and shorter preprocessing time. When point clouds are downsampled, it results in a loss of information. The error in Figure 3.8 is measured as

$$
\left\|o^{\text {est }}-o^{\text {true }}\right\|_{2}+\left\|v_{y}^{\text {est }}-v_{y}^{\text {true }}\right\|_{2}+\left\|v_{x}^{\text {est }}-v_{x}^{\text {true }}\right\|_{2}
$$

with the estimated origin $o^{\text {est }} \in \mathbb{R}^{3}$, the estimated $y$ - and $x$-axis $v_{y}^{\text {est }}$ and $v_{x}^{\text {est }}$ in $S^{2}$, and their ground truth counterparts $o^{\text {true }}, v_{y}^{\text {true }}$, and $v_{x}^{\text {true }}$. We see, that the error slightly increases for increased voxel size. The error is only measured if the algorithm finds two orthogonal
edges. The success rate indicates how often two orthogonal edges could be found. For a voxel size of about 0.020 and more, the algorithm does not find two edges in all trials anymore. Similar results are found for other point clouds of the truck. To minimize the running time of the algorithm while maintaining robust results, we choose a voxel size of 0.018 for the other testing trials. The optimal values for this and the other parameters depend on the way, the investigated point cloud is recorded and generated. LiDAR point clouds, for example, probably need a different voxel size.


Figure 3.8.: Running time, error, and success rate for different voxel sizes for 100 test trials.

Figure 3.9 shows the importance of restricting the edge search to a certain area in the point cloud. The search width is the width of the strip around the estimated plane the loading platform lies in, where we search for the loading edge. This plane is computed with the estimated height and angle of the camera and is therefore not exact. Hence, if the search width is too small, regions containing important information may be removed from the point cloud, leading to a low success rate and high errors. If the search width is too high, the investigated point cloud will contain too many potential edge points leading to unpredictable behaviour in detecting two orthogonal edges.

The uniform downsampling step before computing the edge points has a great influence on the running time of the algorithm. Figure 3.10 shows how the running time changes with different uniform downsampling factors. Around a downsampling factor of 50 , the first signs of unwanted behavior of the algorithm appear. For point clouds with more noise, this can happen more intensively. Throughout the few available testing point clouds, a downsampling factor of 40 has proven to be effective.


Figure 3.9.: Running time, error, and success rate for different search widths for 100 test trials.


Figure 3.10.: Running time, error, and success rate for different uniform downsampling factors for 100 test trials.

### 3.2. Part-Based Pose Estimation Using Factor Graphs

The second and central problem of this work deals with the location of a sensor throughout the entire process of automated loading of a truck. The truck is viewed as an object composed of some simpler objects like the wheels of the truck. Initially, the truck's configuration is only roughly known. Throughout the loading process and with every analyzed detection, this model of the truck is updated and improved such that the sensor pose can be estimated even if only a small part of the truck is visible to the sensors. Figure 3.11 displays this principle of collecting information from far away such that robust pose estimation is possible when the sensor is close to the truck.


Figure 3.11.: Depending on the position of the forklift, the sensors observe different sections of the truck.

To estimate the truck configuration and the sensor pose simultaneously, this structure of the truck as a composite object is modeled as a factor graph as described in Section 2.2. The parts of the truck as well as the sensor in different time steps are represented by variables. Approximate relations between the various parts translate to factors between the respective variables and the observations of the parts of the truck in the sensor frame are expressed as factors between the sensor and the corresponding parts. By optimizing the factor graph as seen in Section 2.3.5, the model is updated and the pose of the sensor in the world frame is estimated. The origin of the world coordinate frame is defined by the beginning of the left loading edge and the axes by the loading edge, the edge between the loading platform and the rear wall, and the vertical direction. Figure 3.12 shows this world frame. Section 3.2.1 describes this translation to a factor graph in more detail.

This process is realized in an algorithm and tested on data obtained from videos of the truck filmed by sensors attached to a forklift. The tool for detecting the parts of the truck correctly is still in development, so the parts of the truck were manually annotated. We use the library GTSAM [11] for factor graph modeling as it provides good ways of defining and using factor graphs, particularly, it is equipped with efficient algorithms for optimizing factor graphs. Section 3.2.1 also gives a brief introduction to the functionality and usage of GTSAM and Section 3.2.2 describes the algorithm of our solution of the part-based pose estimation problem. Finally, in Section 3.2.3, we analyze the algorithm and some parameters. In Appendix B the important parts of the code of our solution can be found.


Figure 3.12.: World coordinate frame for an instance of a truck.

### 3.2.1. Composite Object as a Factor Graph and GTSAM

Factor graphs as described in Section 2.2 are an elegant way of representing a composite object. As they are used in the pose estimation problem, they combine the geometrical structure of the considered object with probability theoretical relations. The Python and C++ library GTSAM [11] (Georgia Tech Smoothing and Mapping) provides data structures and functions for representing and optimizing factor graphs in a variety of applications.


Figure 3.13.: Factor graph representing the truck as a composition of wheels, lights, and loading edges.

We view the truck that can be seen in Figure 3.11 as a composition of its six wheels, two front lights, two rear lights, and the loading edges on both sides. In the code and the following, we refer to the three wheels on the left side of the truck as $w_{i}$, for $i \in\{1,2,3\}$, starting from the front wheel $w_{1}$ back to the last wheel $w_{3}$. On the right side, the wheels are denoted with $W_{1}, W_{2}$, and $W_{3}$. The front lights are named $l_{1}$ and $L_{1}$ and the rear lights $l_{2}$ and $L_{2}$, where the lowercase letters denote the lights on the left side of the truck, and the uppercase letters their counterparts on the right side. The (beginning of the) left
and right loading edges are symbolized as $g_{0}$ and $G_{0}$. The beginning of the left loading edge denotes the origin of the world/global frame. These names can be realized in GTSAM with symbols consisting of one letter and one number each.

In our implemented solution of the pose estimation problem, we create a factor graph at the beginning, representing the geometrical relations between the parts described above. Figure 3.13 visualizes this factor graph. The round vertices depict the variables of the factor graph that represent the parts of the truck. The factor nodes are symbolized by the black squares that lie between pairs of variables. We do not connect each possible pair of two variables, but just those that are somehow close to each other and/or related. For example, the left front wheel $w_{1}$ is connected to its counterpart $W_{1}$ on the other side of the truck, to the left front light $l_{1}$, to the left loading edge $g_{0}$ and to the wheel $w_{2}$.

We realize the factors in GTSAM as BetweenFactorPose3-factors that can be defined by the transformation $T \in \mathrm{SE}_{3}$ that relates the poses $P_{p_{1}}$ and $P_{p_{2}}$ of the parts $p_{1}$ and $p_{2}$ of the truck as $T\left(P_{p_{1}}\right)=P_{p_{2}}$. When viewing the wheels of the truck as right circular cylinders, the class of proper rigid transformations $\mathcal{T}_{P_{w}}^{P_{w^{\prime}}}$ for two wheels $w$ and $w^{\prime}$ of Definition 2.1.17 contains more than one element of $\mathrm{SE}_{3}$. Here, we ignore this problem of symmetry that would result in transformations $T, T^{\prime} \in \mathrm{SE}_{3}$ with $T \neq T^{\prime}$ but $T \sim_{P_{1}} T^{\prime}$ according to Definition 2.1.16, i.e. $T$ and $T^{\prime}$ both could define the factor possibly causing problems when optimizing the factor graph since GTSAM has no functionality for handling symmetries properly. This problem is considered in a different part of the project at AIT that is not regarded in this work. Here, we assume that each object is given by one point in $\mathbb{R}^{3}$ (the centers of the base circles of the wheels viewed as right circular cylinders, the centers of the lights, and the beginnings of the loading edges), defining the position of this object, and a coordinate frame, defining the orientation. Moreover, the notation $T\left(P_{p_{1}}\right)=P_{p_{2}}$ would not be well-defined otherwise, since $P_{p_{1}}$ and $P_{p_{2}}$ denote elements of the pose spaces of objects $p_{1}$ and $p_{2}$ that can only be compared by a transformation $T \in \mathrm{SE}_{3}$ if $p_{1}$ and $p_{2}$ are the same geometric object.

Until now, we just use relative poses between different parts. Thus, the truck could be anywhere in the world frame. But since we define the origin of the world frame at the beginning of the left loading edge, the pose of the whole truck in the world frame cannot be arbitrary. To solve this, we add a factor that is just connected to $g_{0}$. In GTSAM this is called PriorFactorPose3 and we initialize this factor with $(R, t) \in \mathrm{SE}_{3}$ with $t=(0,0,0)^{T} \in \mathbb{R}^{3}$ and $R=I_{3} \in \mathrm{SO}_{3}$.

Later, the sensors start moving around the scene and observing some parts of the truck. For each time step $t$, a new variable $s_{t}$ is introduced, representing the pose of the sensor at that time step. The observations $o_{p}^{(t)}$ in the sensor coordinate frame at time step $t$ define the factor nodes between the sensor $s_{t}$ and the respective part $p$ of the truck. These factors are also realized with BetweenFactorPose3 in GTSAM. After three time steps, the factor graph could look like in Figure 3.14.

Roughly speaking, we start from a model of the truck that defines the factor nodes in the factor graph $\mathcal{F}_{0}$ in Figure 3.13. In each time step $t$, a new sensor node $s_{t}$ is introduced with the observations as factors. The new factor graph $\mathcal{F}_{t}$ is then optimized as illustrated in Section 2.3.5. The factors $f_{j}$ of $\mathcal{F}_{t}$ are defined as probability densities as in (2.22). For more detailed information on how GTSAM defines the factors, see the documentation [12].


Figure 3.14.: Factor graph $\mathcal{F}_{3}$ for the parts of the truck after 3 time steps.

Subsequently, the model is updated according to the outcome of the optimization, i.e. the poses of the parts of the truck in the world frame are being updated as described in Section 3.2.2 in more detail. The model of the truck includes on one hand for each part the estimated pose in the world frame represented as an element of $\mathrm{SE}_{3}$. On the other hand, some dependencies between the different parts describe the configuration of the truck in more detail. For instance, the wheels $w_{1}$ and $W_{1}$ have the same $x$-coordinate, the three wheels on the left, respectively right side of the truck have (approximately) the same $y$ coordinate, and all six wheels have the same height ( $z$-coordinate). These special properties of the truck have to be taken into account when updating the model. Furthermore, some parts can have certain degrees of freedom regarding their pose: The steering angle of the front wheels is not fixed, they can rotate around a vertical axis. Furthermore, the height of the loading edge above the ground is dependent on the weight of the load that is placed on the loading platform (and typically changes throughout the loading process). These features describe the truck (for our purposes) sufficiently well.

### 3.2.2. Description of the Algorithm

Our solution to the pose estimation problem works with one class, keeping track of the currently assumed model of the truck and the factor graph that is enlarged with every time step. The code could be adapted to fit various other composite objects. For using this code on other composite objects, the initial model, defined via the symbols of the different parts, approximate poses of the parts in some world frame, and other object-specific settings, have to be adapted.

The program starts with initializing the factor graph of the truck as well as some other attributes. In each time step, we follow the same three phases. In the first phase, we generate the data. The data consists of the poses of the observed parts. The loading edge can be detected by the loading edge detector described in Section 3.1. The detection of the
other parts is not part of this work. The parts are now manually annotated and will be automatically detected by a machine-learning approach once the whole project is finished. The generated data is handed over to the pose estimator.

The second phase is the pose estimation itself. In particular, the pose of the sensor in the world frame is estimated from the observations of the current time step, previous observations, and the estimated model of the truck. Thus, the sensor tries to locate itself relative to the truck. This is done with the factor graph approach described in Section 3.2.1. The function estimate_sensor_pose() starts by defining initial guesses for the poses of all variables in the factor graph to get a reasonable starting point for the optimization. Using good starting points increases the convergence speed and results in a higher chance of reaching a global optimum. The poses of the variables of the different parts are initialized by their estimated pose in the world frame, given by the current model of the truck. The sensor symbols from previous time steps remain for a certain amount of time steps in the factor graph. Now, old sensor symbols are either deleted from the factor graph to keep the factor graph efficient, or their initial guess is set to their estimated pose from the last time step. The factor graph is expanded by the sensor variable $s_{t}$ of the current time step. The initial guess for the pose of $s_{t}$ is either set randomly, if we are in the first time step, or the pose is initialized by the estimated sensor pose of the last time step. Considering prior knowledge about the starting point of the sensors or odometry information throughout the loading process would give better initial guesses. Subsequently, we optimize the factor graph with the GTSAM version of the Levenberg-Marquardt algorithm as explained in Section 2.3.5.

The final phase of one step of the pose estimation algorithm consists of the model update. The method update_truck_configuration() first updates the poses of all parts and all variables $s_{t^{\prime}}$ for $t^{\prime}<t_{0}$ with the current time step $t_{0}$ that are not deleted from the graph. For a part $p$ with the old pose $T_{p}^{\text {old }}=\left(R_{p}^{\text {old }}, t_{p}^{\text {old }}\right) \in \mathrm{SE}_{3}$, the translation $t_{p}^{\text {old }}$ and the rotation $R_{p}^{\text {old }}$ are updated separately. Let $t_{p}^{\text {old }}$ be the old position in the model, $t_{p}^{\text {est }}$ the position that was estimated in phase 2 of this time step, and $t_{p}^{\text {new }}$ the position that will be the new position of $p$ in the model after this time step. Then we define

$$
\begin{equation*}
t_{p}^{\mathrm{new}}:=w t_{p}^{\text {est }}+(1-w) t_{p}^{\text {old }} \tag{3.2}
\end{equation*}
$$

for some weight $w \in[0,1]$. We take the weighted average of the old and the current estimate of $t_{p}$ to consider the estimated or assumed model of the truck as well as the new estimation. The estimation is based on observations that are noisy due to the restricted capabilities of the sensors and errors in the detection and preprocessing algorithms. Thus, the estimated position $t_{p}^{\text {est }}$ of $p$ is in general not equal to the true position $t_{p}$ and is possibly even further away from $t_{p}$ than $t_{p}^{\text {old }}$. On the other hand, the model of the truck, and therefore $t_{p}^{\text {old }}$, was initialized with a very rough approximation of the true, unknown configuration of the truck, and then updated with noisy data. So the two positions are averaged by (3.2) to hopefully compensate for each other's errors to a certain degree.

The weighting factor $w$ is reduced over time. In the beginning, the model of the truck is assumed to be quite imprecise, so new estimations are weighted more. Throughout the loading process, the truck is observed a lot of times, so the model of the truck gets better over time. Therefore, the current model of the truck gets more weight compared to new
estimations. The weight gets reduced by the function

$$
\begin{equation*}
w(n)=\lambda_{w}^{n}\left(w_{0}-c\right)+c, \tag{3.3}
\end{equation*}
$$

where $w(n)$ is the value of the weight after $n$ time steps. The value $w_{0} \in(0,1]$ is the starting value of the function, $\lambda_{w}$ is the reducing factor in $(0,1)$, and $c \in\left[0, w_{0}\right)$ the lower bound of $w$. According to the computation

$$
\begin{aligned}
w(n) & =\lambda_{w}^{n}\left(w_{0}-c\right)+c \\
\lambda_{w} w(n) & =\lambda_{w}^{n+1}\left(w_{0}-c\right)+\lambda_{w} c \\
\lambda_{w}^{n+1}\left(w_{0}-c\right) & =\lambda_{w} w(n)-\lambda_{w} c
\end{aligned}
$$

the weight $w(n+1)$ can be computed from $w(n)$ as

$$
w(n+1)=\lambda_{w} w(n)+\left(1-\lambda_{w}\right) c
$$

The influence of this factor $\lambda_{w}$ is displayed in Figure 3.18b.
Updating the rotation $R_{p}^{\text {old }}$ is more complicated.
Remark 3.2.1. In [5, Section 7], Brégier et al. describe a method to average two rotations in $\mathrm{SO}_{3}$. First, we take the naive weighted average

$$
Q:=w R_{p}^{\text {est }}+(1-w) R_{p}^{\text {old }}
$$

Then, we compute a singular value decomposition $Q=U D V^{T}$ with $U, V \in \mathrm{O}_{3}$ and the diagonal matrix $D=\operatorname{diag}\left(\alpha_{1}, \alpha_{2}, \alpha_{3}\right)$ with $\alpha_{1} \geq \alpha_{2} \geq \alpha_{3} \geq 0$. The averaged rotation is then given by $R_{p}^{\text {new }}=U S V^{T}$ with $S=\operatorname{diag}(1,1, \operatorname{sgn}(\operatorname{det}(U) \operatorname{det}(V)))$. The matrix $R_{p}^{\text {new }}$ is indeed in $\mathrm{SO}_{3}$.

The new poses $T_{p}^{\text {new }}=\left(R_{p}^{\text {new }}, t_{p}^{\text {new }}\right)$ for all parts $p$ are saved in the model. Special relations between certain parts, as outlined in Section 3.2.1, are considered in two ways. First, GTSAM provides options to equip factors with noise models (e.g. in Section 2.3.5, we modeled the noise of some estimation function $h$ to measurements or assumed relations as normally distributed around 0 with the covariance matrix $\Sigma$ ). The noise for respective coordinates of the construction factors between dependent parts of the truck, e.g. the noise for the $x$-coordinate of the factor between $w_{1}$ and $W_{1}$, is set to a small value by adapting the covariance matrix $\Sigma$ of the noise accordingly. Therefore, when optimizing the factor graph, these relations are valued more than other factors like noisy observations since the probability of lying far from the measurement $z$ is low according to the probability density functions of these construction factors defined via $\Sigma$. Nevertheless, this does not give perfect alignments of dependent parts, so they have to be aligned explicitly every few time steps. The poses of all parts of one dependency group are averaged as described above, regarding their dependent coordinate to obtain a valid model of the truck.

Finally, the model is updated and we are ready for the next time step $t+1$, where we start by updating the factors in the factor graph $\mathcal{F}_{t}$ according to the new model of the truck to get the factor graph $\mathcal{F}_{t+1}$. With $\mathcal{F}_{t+1}$, we continue as described above.

### 3.2.3. Parametrization and Analysis of the Algorithm

The proposed algorithm for the pose estimation of a composite object works in real-time, due to efficient optimization algorithms for factor graphs provided by GTSAM. It is designed to continually receive new observations from the sensors to improve the estimated model of the truck and estimate the pose of the sensors relative to the truck to enable autonomous loading. A limiting factor for realizing this approach is the correct detection and classification of the different parts of the truck. The respective machine-learning tool, that is planned to be employed in this project, is still in development. Since time is not a crucial aspect of this approach, we want to analyze the influence of different parameters on the outcome of the model.


Figure 3.15.: Estimated sensor trajectories in four different phases: The yellow cubes indicate the estimated sensor positions and the small red cubes indicate the true sensor positions.

As stated above, there are still missing parts in the whole project of automated loading of a truck. Therefore, it was not possible to generate the data under conditions that replicate those present in a real-world deployment of the pose estimator. With a ZED Camera attached to a forklift, we recorded four sequences of a truck while driving around the truck. The blender model of the truck was then fitted with GeoTracker for Blender from KeenTools [35] into the scene and tracked throughout the recordings. We define a
time step as one second. For each time step, the poses of the different parts of the truck that were visible for the sensor (i.e. within the field of view and not hidden by the rest of the truck) in the respective sensor coordinate frames were extracted and used as the testing data. Figure 3.15 shows the estimated trajectories of the four different loading scenarios. In Phases II and IÍI, the truck was approached once from the back and once from the front. In Phase I, the truck was viewed from the other side, and Phase IV simulated a loading process.

But first, we want to see how our pose estimation approach updates the model of the truck. Figure 3.16 shows some time steps in the pose estimation process. We start with a very rough approximation of the model as displayed in Figure 3.16a. The reddish truck pictures the assumed ground truth of the model, the blue cylinders portray the estimated (or initial) poses of the wheels, the blue cuboids portray the estimated (or initial) poses of the front and the rear lights, and the green cuboids picture the estimated (or initial) poses of the loading platform and the driver's cabin (for illustration purposes). The three arrows at the beginning of the left loading edge visualize the origin of the world coordinate frame. The other three arrows show the estimated pose of the sensor at that time step. As usual, the red arrow denotes the $x$-axis, the green arrow denotes the $y$-axis, and the blue arrow denotes the $z$-axis. Since the poses of the parts were annotated with Blender [9], the camera looks in the negative $z$-direction. At time step 5 in 3.16 c , the rear wheels and lights are already approximately at the correct positions. Their positions become more accurate when the sensor detects the rear lights again in Phase II as seen in 3.16e. In Phase III, the front lights are observed for the first time, thus, Figure 3.16 f already provides a good model.

We analyze several different parameters and their influence on the model and the estimation of the sensor pose. To measure the performance of the algorithm with a certain parameter set, we use the metric $d$ on $\mathrm{SE}_{3}$ defined in Theorem 2.1.31. We define the cumulated model error $\operatorname{err}^{(t)}$ in time step $t$ as

$$
\operatorname{err}^{(t)}:=\sum_{p \in P} d\left(T_{p}^{(t)}, T_{p}^{\text {true }}\right)
$$

where $P$ denotes the set of all parts of the truck, $T_{p}^{(t)} \in \mathrm{SE}_{3}$ defines the pose of part $p \in P$ at time step $t$, and $T_{p}^{\text {true }}$ defines the pose of the part $p$ in the assumed ground truth. We take a look at the initial weights for updating the translation and rotation of a part, at the construction and observation noise, and at the factors for reducing the construction noise and the weights. For further descriptions of these parameters, see Section 3.2.2.

For each parameter $\lambda$ at a time, we analyze the cumulated model error err ${ }^{(t)}$ for the test data described above by setting $\lambda$ to different values, assuming the rough initial model as in the first picture of Figure 3.16, and passing the data to the pose estimator multiple times, each time with some additional random noise on the initial model and the observations. For the initial model, we add random numbers of the normal distribution $\mathcal{N}\left(0,0.5^{2}\right)$ to certain distances of parts of the model (e.g. the distance between wheels $w_{1}$ and $w_{2}$ ). To add noise to the rotations, we take a vector $\xi \in \mathbb{R}^{3}$ with entries sampled from $\mathcal{N}\left(0,0.02^{2}\right)$ and update the rotation $R$ with the local update $\mathcal{R} \oplus \xi$ of Definition 2.3.6. The standard deviation for the rotation is rather small since the parts of trucks have fixed rotations to


Figure 3.16.: Estimated model (green and blue) and ground truth (reddish) of the truck for a pose estimation run of Phases I to IV.
each other, regardless of the specific truck configuration, e.g. the two loading edges are always parallel and the lights are always oriented directly to the front and the back. The observations are treated similarly but with a standard deviation of 0.1 on the translation and 0.05 on the rotation.

We start with the different noise models that are used here. In Section 2.3.5 it is stated that the probability density of a factor in a factor graph can be interpreted as a noise model.


Figure 3.17.: Model and sensor estimation error depending on the noise for the translation of the construction factors.

(a) Model error depending on the updating factor $\lambda_{c}$ for the noise of the construction factors.

(b) Model error depending on the updating factor $\lambda_{w}$ for the weight $w$ of model updates.

Figure 3.18.: Model error for different values of the updating factors $\lambda_{c}$ and $\lambda_{w}$.

In GTSAM, each factor is initialized with a noise model. In the created factor graph (see Figure 3.14), we distinguish among the construction factors between two different parts of the truck, that are defined by the currently estimated model, and the observation factors $o_{p}^{(t)}$ defined by the observations of different parts.

The construction factors express in the beginning a rough model with additional noise. Thus, we need higher values for the construction noise than the standard deviation of the additional noise. Figure 3.17a displays the model error for a few different values of the construction noise. The Regions I, II, III, and IV indicate the four different sequences of the sensor movement around the truck as seen in Figure 3.15. First, the sensor approaches the truck from the back right, then from the back left, then from the front left, and in Phase IV, the forklift simulates the loading of a palette. This is a reasonable real-life scenario since the forklift moves around near the truck before it starts loading stuff. Thus, the model must be quite accurate in Phase IV, because when the forklift loads palettes onto the truck, the sensors are too close to detect many parts.

For each parameter value, the computation was simulated 50 times. The lines display the arithmetic mean of the model errors for a certain parameter value. The shaded areas around the lines indicate the corresponding empirical standard deviations. We see that if the construction noise is too small, the factor graph relies too much on wrong assumptions and fails to optimize the model properly. If the construction noise is too high towards the end, the model error rises again since the already well-adapted model is loaded with too much uncertainty and new noisy observations make the model worse.

Figure 3.17 b was created for the same situation but displays the error of the sensor estimation measured as $d\left(T_{s_{t}}^{\text {est }}, T_{s_{t}}^{\text {true }}\right)$ with the metric $d$ on $\mathrm{SE}_{3}$ of Theorem 2.1.31, the estimated sensor pose $T_{s_{t}}^{\text {est }}$ at time step $t$, and the corresponding ground truth $T_{s_{t}}^{\text {true }}$. This gives similar results on good values for the construction noise. In Region IV, around the timesteps 105 to 110 , we see a short rise in the estimation error. At that time, the forklift placed the load on the truck and detected almost no parts of the truck.

These considerations indicate that the noise for the construction factors should change over time. In the beginning, a high construction noise ensures a flexible model. Towards the end, we already have a good approximation of the model, so a low construction noise ensures that we consider this model enough. Therefore, we introduce an updating parameter $\lambda_{c}$ for the construction noise similar to the updating parameter $\lambda_{w}$ in (3.3) and analyze $\lambda_{c}$ in Figure 3.18a. If this factor is too low, the construction noise gets too small too fast, and the model does not update properly anymore. If $\lambda_{c}$ is too high or even set to 1 on the other hand, the effect of a rising model error towards the end, as described above, occurs. For the updating factor $\lambda_{w}$ of the weights for updating the model, we observe similar results as displayed in Figure 3.18b.

We set the standard deviation of the artificial noise on the translation part of the observations to 0.1 . Considering the noise occurring in the data generation process, this noise is probably a bit higher than 0.1. Indeed, as seen in Figure 3.19, for a value of 0.17 , the model gets updated optimal. If the observation noise is too low, the noisy observations are considered too much. If the observation noise is too high, at some point, the model cannot be improved anymore.


Figure 3.19.: Model error depending on the noise for the translation of the observation factors.

### 3.3. Possible Improvements and Further Work

There are several directions, that go beyond the scope of this master's thesis, to extend the described workflow (especially the pose estimator). The greatest part is the integration of the loading edge detector and the pose estimator in the bigger project at AIT. Once the relevant tools, such as the machine-learning tool for detecting and classifying various parts, are completed, the code and parameters must be adjusted to meet the requirements of the new data and any downstream algorithms that rely on the pose estimator.

Furthermore, the pose estimator could be adapted to better fit real-life situations. For instance, we could introduce and test objects with certain degrees of freedom. The orientation of the front wheels depends on the current steering angle and is therefore flexible throughout the algorithm. Additionally, the height of the loading platform depends on the weight of the load placed on it. Degrees of freedom could be realized by special factors or noise models that express these specific properties. Likewise, the algorithm could be adapted to accept incomplete poses, e.g. the loading edge is correctly detected, but the beginning of the loading edge cannot be determined. Similar to poses with degrees of freedom, this could be handled by adapting the observation noise in the corresponding directions. Considering the short rise of the estimation error of the sensor pose when the load is being placed on the truck (as seen between time steps 105 and 110 in Figure 3.17b) one might pose the question of how to handle the situation if only little or no parts of the truck can be detected. Here, we cut the parts of the video, where no parts of the truck
were observed. In real applications, a system switching between this pose estimator and other methods of navigating around the truck could be employed. For instance, GPS and odometry measurements could be taken into account in the pose estimator as well as in other methods. With all these improvement ideas, one has to be careful to maintain an efficient algorithm.

With tools that generate data for this algorithm automatically, we are faced with the problem of incorrect data. For example, the classification of the wheels could be mixed up, mislabeling the front wheel as $w_{3}$ or a wheel on the left side as $W_{2}$. Moreover, one has to be prepared for a completely wrong detection, where the data generating tool detects a street lamp as some light of the truck or some circular load as a wheel of the truck. Since the factor graph already carries a probability structure, we could estimate how likely certain detections of different parts of the truck are, and consequently discard wrong detections before estimating a new model. More comprehensive testing of various situations and different trucks is necessary to ensure a responsible and practical implementation of this technology in real life.

Lastly, one could test the concept of detection of composite objects on objects other than the truck. A bicycle is composed of two wheels (one with a certain degree of freedom), a handlebar, a saddle, a frame (consisting of cylinder-like shapes), pedals, lights, a chain, and so on. Depending on the use case and available sensors, one could go into detail or stick to fewer and simpler objects. A robot arm loading different kinds of bicycles in some compartments might need to detect bicycles without knowing their specific configurations.

Thus, there are various related problems and use cases this work could be extended to.

## 4. Conclusion

In this thesis, we developed and analyzed two steps of an automated truck-loading process and examined the mathematical foundations of the used methods.

The loading edge detection algorithm is based on the geometric properties of points in point clouds. First, the point cloud is downsampled with voxel downsampling and outlier removal. This downsampling process is a tradeoff between the efficiency and robustness of the algorithm. For a small voxel size, the downsampling of the point cloud and all steps of the algorithm that are executed on all points in the corresponding point cloud take a lot of time. If the voxel size is too big, too much information about the truck gets lost and the error of the detected loading edge to the ground truth rises, while the success rate of the algorithm shrinks.

Then, the resulting point cloud is cut to the strip of points containing the loading platform, using the estimated height and angle of the camera. A test on the width of this strip showed the importance of this cutting step, and therefore the importance of accurate estimates of the height and angle of the camera in this approach. The camera sensors produce noisy data, forcing us to relax the conditions in the search for edge points. Subsequently, the risk of declaring points as edge points incorrectly rises. So, if the search width is smaller, more edge-like structures in the point cloud are cut away.

The edge point detection is then executed on every $n$-th point in the remaining strip. This random downsampling factor $n$ has proven to have a great influence on the running time of the algorithm. In real-life applications, this parameter has to be adapted to the density of the point cloud (depending on the sensors and the previous downsampling steps). Some testing under the actual circumstances is necessary to find the range of this parameter, where the algorithm runs efficiently, but the success rate stays high and the error stays low. This loading edge detection algorithm can be used in the data generation process for the pose estimator.

The pose estimator takes observations of some parts of the truck and tries to reconstruct the truck's configuration from a rough initial model as well as the pose of the sensor relative to the truck. Therefore, we considered the truck as a composite object, composed of its wheels, lights, and loading edges, and represented this structure as a factor graph. A rough initial approximation of the spatial relations of the different parts of the truck to each other translated to the construction factors between the respective variables.

The spatial relations were given as rigid transformations in $\mathrm{SE}_{3}$. We have seen that $\mathrm{SO}_{3}$ and $\mathrm{SE}_{3}$ are smooth manifolds, allowing the use of manifold optimization techniques in this pose estimation problem. The retraction for these manifolds was defined via the exponential map for matrices. The power series of the exponential map has for skew-symmetric matrices in $\mathbb{R}^{3 \times 3}$ an explicit representation as Rodrigues' formula. For manifold optimization problems on a higher dimensional manifold $\mathrm{SO}_{n}$ for $n>3$, either an approximation of the exponential map has to be used or different retractions have to be considered to obtain
efficient optimization algorithms.
We visualized the updating process of the model. After a small number of time steps, the estimated poses of the observed parts got close to the ground truth. To model certain dependencies of parts on each other, the probability densities of the respective factors have to be adapted accordingly. Due to this functionality, the estimations of the poses of the wheels of the truck on both sides improved even while the camera stayed on one side of the truck observing only the right wheels.

The uncertainty of the model was implemented via the construction noise. Tests on this parameter showed that the construction noise has to be high enough to reflect the uncertainties of the construction factors. Especially at the beginning of the pose estimation process, the model is only a rough estimation of the real truck configuration. The model of the truck improves with every step, therefore, the construction noise needs to be carefully reduced throughout the loading process. If the construction noise is reduced too quickly or too slowly, we receive high model errors towards the end of the optimization process as the respective test showed.

The observations of the parts of the truck are represented by the observation factors in the factor graph. The corresponding observation noise should reflect the errors of the sensors and the preprocessing algorithms. For the construction and the observation noise, more testing on real data in various scenarios is required. The estimation of the sensor pose depends on the accuracy of the model and the number and quality of observations. We have seen that during the actual loading process, the sensors get so close to the truck that little to no parts can be observed. This leads to high sensor pose estimation errors during this phase. For real-life applications, different methods or additional sensors pointing in different directions have to be considered during the loading phase.

We presented a new application of factor graph modeling. The truck serves as an example of this composite object detection approach. For a full proof of concept, more tests on different composite objects have to be made.

## A. Code Loading Edge Detection

```
import numpy as np
import open3d as o3d
import time
```

```
def run_loading_edge_detection(filename, inputs):
    """"
    Run the loading edge detection.
    Parameters
    filename : str
        Name of the file containing the point cloud.
    inputs : dict
        Dictionary containing some custom parameter values.
    Returns
    loading_edge : ndarray of shape (2,3)
            Two points that define the loading edge.
    other_edge : ndarray of shape (2,3)
            Two points that define the second edge.
    running_time : float
        Duration of the loading edge detection.
    num_edges_found : int
        0, 1, or 2, dependent on the number of edges found.
    """
    # read in the point cloud
    pcd = o3d.io.read_point_cloud(filename)
    # define an instance of the class with the desired parameters
    led = LoadingEdgeDetection(o3d_cloud=pcd, **inputs)
    # perform voxel downsampling
    voxel_downsampled_pcd = led.voxel_downsample_pcd(led.pcd, voxel_size=0.015)
    # remove the outliers
    final_pcd = led.remove_outliers(voxel_downsampled_pcd, nb_points=30, radius=0.05)
    # find the loading edge
    loading_edge, other_edge, running_time, num_edges_found = led.find_two_edges(final_pcd)
    return loading_edge, other_edge, running_time, num_edges_found
```

class LoadingEdgeDetection:
def __init__(
self,
o3d_cloud,
est_angle_camera=32.5,
est_height_camera=2.05,
est_height_loading_platform=1.35,
search_width=0.5,
tol_orthogonal=0.3,
max_lines=5
uniform_downsample_factor=40,

Constructor.

Parameters
o3d_cloud : open3d PointCloud
Point cloud, where the loading edge should be detected.
est_camera_angle : float, optional
Estimated camera angle in degree to the horizontal plane. The default is 32.5.
est_height_camera : float, optional
Estimated height of the camera above the ground in meters. The default is 2.05.
est_height_loading_platform : float, optional
Estimated height of the loading platform above the ground in meters. The default is 1.35.
search_width : float, optional
Width of the remaining point cloud, when cutting the pcd above and below
the estimated plane of the loading platform. The default is 0.5.
tol_orthogonal : float, optional
Tolerance when searching for orthogonal lines. Has to be between 0 and 1.
The default is 0.3.
max_lines : int, optional
Maximum number of lines to be found in 'line_ransac'. The default is 5.
uniform_downsample_factor : int, optional
Downsample factor when reducing the number of points for the edge point search.
The default is 40 .

Attributes
pcd : open3d PointCloud
Original point cloud of the scene.
points : ndarray of shape (num_points, num_dimensions)
Points of the original point cloud as numpy array.
est_height_camera : float
Estimated height of the camera above the ground in meters. Used to estimate the pose of the loading platform.
est_height_loading_platform : float
Estimated height of the loading platform above the ground in meters. Used to estimate
the pose of the loading platform.
up_vector : ndarray of shape (3,)
Estimated up-vector of the scenery. Computed with the estimated angle of the camera.
point_loading_platform_below_camera : ndarray of shape (3,)
Point in the estimated height of the loading platform, vertically below the camera.
Computed by the estimated up-vector and height of the camera and the loading platform.
search_width : float
Width of the remaining point cloud, when cutting the pcd above and below the estimated
plane of the loading platform. Smaller search width speeds up the computation and
eliminates other edge points. Can be decreased, if angle and height estimates are good.
tol_orthogonal : float in [0, 1]
Tolerance when searching for orthogonal lines. It holds:
$v$ is orthogonal to $w \ll \quad$ @ $w=0$.
Normalized vectors $v$ and $w$ are approximately orthogonal, if $v$ @ $w<$ tol_orthogonal.
max_lines : int
Maximum number of lines to be found in 'line_ransac' before the algorithm stops.
Larger max_lines is more likely to find a second edge,
but also more likely to declare a line as an edge, that is not an edge.
origin : ndarray of shape (3,)
Origin of the world coordinate frame, defined in the method find_origin(). The origin of the world frame is defined as the beginning of the left loading edge. If the truck is seen from the right side, 'origin' states the beginning of the right loading edge.
coordinate_frame : dict
Contains the directions of the coordinate axes of the world frame, defined in the method

```
    rotate_coordinate_frame(). The keys are ' }x\mathrm{ ', ' 'y', and 'z'.
uniform_downsample_factor : int
    The edge point detection is only performed on some points, to save computation time.
    For a downsampling factor of n, every n-th point is investigated.
"""
# define the point cloud
self.pcd = o3d_cloud
# define the estimated height of the camera and the loading platform
self.est_height_camera = est_height_camera
self.est_height_loading_platform = est_height_loading_platform
# compute the estimated up-vector of the scenery from the estimated camera angle
# and the point below the camera, in the estimated height of the loading platform
self.up_vector = self.get_up_vector(est_angle_camera)
self.point_loading_platform_below_camera = self.get_point_loading_platform_below_camera()
# define additional parameters
self.search_width = search_width
self.tol_orthogonal = tol_orthogonal
self.max_lines = max_lines
self.coordinate_frame = {}
self.uniform_downsample_factor = uniform_downsample_factor
```

\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
\# Point Cloud Processing
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
def cut_pcd_with_plane(self, pcd, weights, dist_to_original_plane=0.5, inside=False):
Cut a point cloud with a plane.
Use this function, to speed up computation by restricting the search space.
Parameters
pcd : open3d PointCloud
Point cloud to be cut.
weights : list of float
Weights $a, b, c, d$ of the $p l a n e: a * x+b * y+c * z+d=0$.
dist_to_original_plane : float, optional
Distance of the planar cut to the original plane. The default is 0.5.
inside : bool, optional
If True, the points inside the cut region are returned.
If False, the points outside the cut region are returned.
The default is False.
Returns
-------
filtered_pcd : open3d PointCloud
Cut point cloud.
" " " "
\# get the indices of the points inside of the cut region
filtered_indices = self.cut_pcd_with_plane_indices(
self.pcd_to_np(pcd), weights, dist_to_original_plane, inside
)
\# select the points with the respective indices
filtered_pcd = pcd.select_by_index(filtered_indices)
return filtered_pcd
def cut_pcd_with_plane_indices(self, pcd_np, weights, dist_to_original_plane, inside):
"""

Computes the indices of the points inside or outside the cut region for cut_pcd_with_plane().

```
    Parameters
```

    \(p c d \_n p\) : ndarray of shape (num_points, num_dimensions)
        Point cloud as numpy array.
    weights : list of float
        Weights \(a, b, c, d\) of the \(p l a n e: ~ a * x+b * y+c * z+d=0\).
    dist_to_original_plane : float
        Distance of the planar cut to the original plane.
    inside: bool
        If True, the points inside the cut region are returned.
        If False, the points outside the cut region are returned.
    Returns
    list of int
        Indices of the points inside or outside the cut region.
    """
    \# transpose the array to ease computation ( \(p c d \_n p[0]\) is a vector of all first coordinates)
    pcd_np \(=\) pcd_np. \(T\)
    \# compute the values of the plane equation for all points as a numpy array
    values \(=\) weights[0] * pcd_np[0] + weights[1] * pcd_np[1] + weights[2] * pcd_np[2] + weights[3]
    \# cut in both directions
    above = values > np.abs(dist_to_original_plane)
    below = values < -np.abs(dist_to_original_plane)
    \# return the indices of the points inside or outside the cut region
    if not inside:
        return [i for \(i\) in range(len(above)) if above[i] or below[i]]
    else:
        return [i for \(i\) in range(len(above)) if not above[i] and not below[i]]
    " " " "
Additional functions, not shown here:
def divide_inlier_outlier(self, pcd, inlier_ind, color)
def voxel_downsample_pcd(self, pcd, voxel_size)
def remove_outliers(self, $\left.p c d, n b \_p o i n t s, ~ r a d i u s\right) ~$
def $p c d_{-}$to_np (self, $p c d$ )
def draw_point_cloud(self, list_of_pcds, show_normals, estimate_normals)
"""
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
\# Truck Specific Functions
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
def get_up_vector(self, camera_angle):
"""
Compute the estimated up-vector (up-direction of the scenery in the world frame) from the
estimated camera angle as np.array ([np. sin(-camera_angle), 0, np.cos(-camera_angle)]).
Parameters
camera_angle : float
Estimated camera angle in degrees to the horizontal plane.
Returns
up_vector: ndarray of shape (3,)
Approximate up-vector.
\# z-axis goes up ( $x$ to the front, $y$ to the side)
up_vector_camera $=n p \cdot \operatorname{array}([0,0,1])$
\# angles to rotate around (rotate around the $y$-axis) in radians
angles $=n p . d e g 2 r a d\left(n p . a r r a y\left(\left[0,-c a m e r a \_a n g l e, ~ 0\right]\right)\right)$
\# rotation matrix
rot_matrix = o3d.geometry.get_rotation_matrix_from_axis_angle(angles)
return rot_matrix @ up_vector_camera
def get_point_loading_platform_below_camera(self):
""""
Computes the point below (in the direction of the estimated up-vector) the camera
that lies in the estimated height of the loading platform.
Returns
-------
point : ndarray of shape (3,)
Estimated position of the point in the height of the loading platform, below the camera.
Given in sensor frame coordinates.
"""
\# the camera defines the origin of the sensor frame
pos_camera $=$ np. $\operatorname{array}([0,0,0])$
\# height difference between camera and loading platform
height_diff_camera_loading_platform = (
self.est_height_camera - self.est_height_loading_platform
)
return pos_camera - height_diff_camera_loading_platform * self.up_vector
def cut_pcd_around_loading_platform(self, pcd):
""""
Cuts a point cloud above and below the estimated loading platform. The normal vector of the
plane, the height of the plane, and the width of the cutted point cloud are set as attributes.
Parameters
pcd : open3d PointCloud
Input point cloud.
Returns
open3d PointCloud
Output point cloud.
" "" "
\# equation of the plane: $n X=n P \rightarrow n X-n P=0$
\# the up-vector defines the normal vector $n$ of the plane
weights $=$ list(self.up_vector)
\# append $-n P$ to the weights
weights.append(-self.up_vector @ self.point_loading_platform_below_camera)
\# cut the point cloud above and below the plane according to the search width
return self.cut_pcd_with_plane(pcd, weights, self.search_width / 2, True)
def find_origin(self, P_line1, P_line2):
"""
Find the origin of the coordinate frame as the intersection point of two given lines.
These lines do not intersect exactly, so we intersect one of the lines with the plane that is
defined by the other line and the vertical direction.

Then we take the average of the two intersection points.

Parameters

P_line1 : ndarray of shape $(2,3)$
Two points that define the first line.
$P_{-}$line2 : ndarray of shape $(2,3)$
Two points that define the second line.
set_origin : bool, optional
If True, the origin of the coordinate frame is set as an attribute of the class. The default is False.

Returns
origin: ndarray of shape (3,)
Found origin of the coordinate frame.
" ""
\# get the intersection points
intersection_point1 = self.intersect_line_plane(
P_line1, np.array([P_line2[0], P_line2[1], P_line2[0] + self.up_vector])
)
intersection_point2 = self.intersect_line_plane(
P_line2, np.array([P_line1[0], P_line1[1], P_line1[0] + self.up_vector])
)
\# compute the origin and set it as a class attribute
origin $=$ (intersection_point1 + intersection_point2) / 2
self.origin = origin
return origin
def find_direction(self, edge_pcd, origin, P):
" " "
This function is used to find the true direction of an axis (deciding between vec and -vec)
to find the correct coordinate frame. The correct direction is the vector from the origin
to the endpoint of $P$ that is further away from the origin. This is done by finding the
endpoints of the line and comparing the distances of these endpoints to the origin.

## Parameters

edge_pcd : open3d PointCloud
Point cloud that contains the points on the edge.
This point cloud should already be the largest cluster of a found edge.
origin : ndarray of shape (3,)
Origin of the coordinate frame.
$P$ : ndarray of shape $(2,3)$
Two points that define the line.
The line defined by $P$ does not have to be parallel to the line defined by the endpoints. The two endpoints might just give a rough approximation of the direction of the line.

## Returns

$P$ : ndarray of shape $(2,3)$
Two points that define the line.
They are the same points as the input $P$, but the order of the points might be switched. """
\# direction vector of the line
$\mathrm{vec}=\mathrm{P}[1]-\mathrm{P}[0]$
\# get the approximate endpoints of the line
endpoint1, endpoint2 = self.find_endpoints(edge_pcd, max_iter=100)
\# compare the distances of the endpoints to the origin

```
dist_origin_endpoint1 = np.linalg.norm(endpoint1 - origin)
dist_origin_endpoint2 = np.linalg.norm(endpoint2 - origin)
if dist_origin_endpoint1 > dist_origin_endpoint2:
    # if endpoint1 is further away from the origin than endpoint2
    # -> the correct approximate direction is the vector from the origin to endpoint1
    correct_direction = endpoint1 - origin
else:
    # otherwise the correct approximate direction is the vector from the origin to endpoint2
    correct_direction = endpoint2 - origin
# compare the correct direction to the original direction
P_start = origin
correct_direction /= np.linalg.norm(correct_direction)
if correct_direction @ vec > 0:
    # if their inner product is positive, the direction is correct
    P_end = P[1]
else:
    # otherwise the direction is wrong and has to be switched
    P_end = P[0]
return np.array([P_start, P_end])
def rotate_coordinate_frame(self, vec_loading_edge, vec_other_edge):
"""
Rotate the right-handed coordinate frame such that the loading edge is the x-axis (left
loading edge) or the y-axis (right loading edge). The vector of the loading edge defines its
axis exactly, the z-axis is then the cross-product of vec_loading_edge and vec_other_edge,
and the last axis is the cross-product of the z-axis and the axis of the loading edge.
We assume, that the coordinate frame is already centered at the correct origin.
Since we transform an orthonormal basis to another orthonormal basis,
the rotation matrix is given by the new axes as columns.
Parameters
vec_loading_edge : ndarray of shape (3,)
    The direction vector of the detected loading edge.
vec_other_edge : ndarray of shape (3,)
    The direction vector of the other edge.
Returns
rot_matrix : ndarray of shape (3,3)
    Rotation matrix to the new coordinate frame.
" """
# check if the loading edge becomes the x- or the y-axis
if self.up_vector @ np.cross(vec_loading_edge, vec_other_edge) > 0:
    # the loading edge is the x-axis
    new_x_axis = vec_loading_edge
    new_z_axis = np.cross(new_x_axis, vec_other_edge)
    new_y_axis = np.cross(new_z_axis, new_x_axis)
else:
    # the loading edge is the y-axis
    new_y_axis = vec_loading_edge
    new_z_axis = np.cross(vec_other_edge, new_y_axis)
    new_x_axis = np.cross(new_y_axis, new_z_axis)
# normalize the new axes to get the rotation matrix
new_x_axis /= np.linalg.norm(new_x_axis)
new_y_axis /= np.linalg.norm(new_y_axis)
new_z_axis /= np.linalg.norm(new_z_axis)
rot_matrix = np.array([new_x_axis, new_y_axis, new_z_axis]).T
```

```
# set the axes and the rotation matrix as attributes of the class
self.coordinate_frame["x"] = new_x_axis
self.coordinate_frame["y"] = new_y_axis
self.coordinate_frame["z"] = new_z_axis
self.rot_matrix_to_coordinate_frame = rot_matrix
return rot_matrix
```

```
##############################################################################################
```

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# Edge Detection

# Edge Detection

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def find_two_edges(self, pcd):
Find two orthogonal edges in a point cloud. The first edge is found by 'find_longest_edge'
Then, a second edge is found that is horizontal and orthogonal to the first edge. If two edges
are found, the function 'differentiate_edges' is used to distinguish the loading edge from
some other edge. If necessary, the two edges are switched. The origin of the coordinate frame
is set as the intersection point of the two edges. The orientation of the coordinate frame is
set such that the loading platform is located in the first quadrant of the xy-plane.
The two edges are then drawn together with the rest of the point cloud.
Parameters
pcd : open3d PointCloud
Point cloud, where the edges should be detected.
Returns
edge1_P : ndarray of shape (2,3)
Two points that define the loading edge.
edge2_P : ndarray of shape (2,3)
Two points that define the second edge.
time : float
Duration of the loading edge detection.
num_edges_found : int
0, 1, or 2, dependent on the number of edges found.
"""
\# start the timer
start = time.time()
\# find the longest edge in the point cloud
edge1_P, edge1_pcd, edge1_ind, edge_points_pcd, found_longest_edge = self.find_longest_edge(
pcd
)
\# if there are enough edge points to find an edge, we look for a second edge
if found_longest_edge:
\# find a second edge that is orthogonal to the first edge
edge2_P, second_line_found = self.find_orthogonal_line(
edge_points_pcd, P_original=edge1_P, max_lines=self.max_lines
)
\# if a second edge is found, the coordinate frame can be set
if second_line_found:
num_edges_found = 2
origin = self.find_origin(edge1_P, edge2_P)
edge2_ind = self.get_points_on_line(pcd, edge2_P)
edge2_pcd, _ = self.divide_inlier_outlier(pcd, edge2_ind, color=[0, 1, 0])
edge1_P = self.find_direction(edge1_pcd, origin, edge1_P)

```
```

edge2_P = self.find_direction(edge2_pcd, origin, edge2_P)

# define the direction of the up-vector (currently it is up or down)

# the origin lies below the camera, so the angle between the up-vector

# and the vector world_origin -> sensor_origin should be smaller than 90'

# if the angle is larger than 90', the up-vector is pointing downwards

if self.up_vector @ (np.array([0, 0, 0]) - origin) < 0:
self.up_vector *= -1

# check if edge1 or edge2 is the loading edge

_, _, correct_order = self.differentiate_edges(pcd, edge1_P, edge2_P)

# switch edge1 and edge2 if necessary

if not correct_order:
edge1_P, edge2_P = edge2_P, edge1_P
edge1_pcd, edge2_pcd = edge2_pcd, edge1_pcd
edge1_ind, edge2_ind = edge2_ind, edge1_ind

# get the rotation matrix to rotate the coordinate frame

rot_matrix, axis_loading_edge = self.rotate_coordinate_frame(
edge1_P[1] - edge1_P[0], edge2_P[1] - edge2_P[0]
)
end = time.time()

# here would be the place to visualize the results

return edge1_P, edge2_P, end - start, num_edges_found
else:
num_edges_found = 1
\# if only one edge is found,
\# the second horizontal direction is determined by the first edge and the up-vector
edge1_P = self.find_endpoints(edge1_pcd)
\# the origin is one endpoint of the first edge
origin = edge1_P[0]
\# get the direction of the second axis
potential_second_axis = np.cross(edge1_P[1] - edge1_P[0], self.up_vector)
\# the loading platform locates on the side of the first edge, where more points are
lifting_factor = 0.3
lift1 = lifting_factor * potential_second_axis
edge1_P_lifted1 = edge1_P + lift1
lift2 = lifting_factor * (-potential_second_axis)
edge1_P_lifted2 = edge1_P + lift2
num_votes_lifted_1 = len(
self.get_points_on_line(
pcd, edge1_P_lifted1, tol=0.1, cluster=False, between_points=True
)
)
num_votes_lifted_2 = len(
self.get_points_on_line(
pcd, edge1_P_lifted2, tol=0.1, cluster=False, between_points=True
)
)
if num_votes_lifted_1 < num_votes_lifted_2:
second_axis = -potential_second_axis
else:
second_axis = potential_second_axis

```
```

second_axis /= np.linalg.norm(second_axis)

# get the rotation matrix to rotate the coordinate frame

rot_matrix = self.rotate_coordinate_frame(edge1_P[1] - edge1_P[0], second_axis)
end = time.time()

# here would be the place to visualize the results

edge2_P = np.array([origin, origin + second_axis])

```
return (edge1_P, edge2_P, end - start, num_edges_found)
def find_longest_edge(self, pcd):
    Find the longest edge in a point cloud. First we find the edge points and then a line that
    contains many of these edge points.
    Parameters
    pcd : open3d PointCloud
        Point cloud, where the edge should be detected.
    Returns
    edge_P : ndarray of shape \((2,3)\)
        The endpoints of the dominant edge in the point cloud.
    edge_pcd : open3d PointCloud
        Point cloud that contains the points on the dominant edge.
    edge_ind: list of int
        Indices of the points on the dominant edge.
    edge_points_pcd : open3d PointCloud
        Point cloud that contains the edge points.
    edge_found : bool
        Indicates if the algorithm found an edge.
    """
edge_found = True
\# cut the point cloud above and below the estimated plane of the loading platform
pcd_cut \(=\) self.cut_pcd_around_loading_platform(pcd)
\# find the edge points with the normals of the points
edge_points_ind = self.find_edge_points_normals(pcd_cut)
\# check if enough edge points were found
if len(edge_points_ind) < 3:
        edge_found = False
        return None, None, None, None, edge_found
    \# divide the cut point cloud into edge points and non-edge points
    edge_points_pcd, _ = self.divide_inlier_outlier(pcd_cut, edge_points_ind)
    \# find a line that contains many of these edge points
    P, ransac_completed = self.line_ransac(edge_points_pcd)
    if ransac_completed:
        \# find the points of the original point cloud on this line
        edge_ind = self.get_points_on_line(pcd, P, between_points=False)
        \# divide the original point cloud into edge points and non-edge points
        edge_pcd, _ = self.divide_inlier_outlier(pcd, edge_ind, color=[1, 0, 0])
```

edge_P = self.find_endpoints(edge_pcd)
return edge_P, edge_pcd, edge_ind, edge_points_pcd, edge_found
else:
edge_found = False
return None, None, None, None, edge_found
def find_edge_points_normals(self, pcd, radius=0.1):
"""
Find edge points of a point cloud by computing the normals of the neighbors of each point.
If the normals can be clustered into two groups reasonably, such that the mean of one group
is nearly parallel to the vertical direction, and the mean of the other group is nearly
orthogonal to the vertical direction, the point is considered an edge point.
Uses the function 'cluster_normals' to cluster the normals.
Parameters
pcd : open3d PointCloud
Point cloud, where the edge points should be found.
radius : float, optional
Radius around a point, where the kd-tree looks for neighbors. The default is 0.1.
Returns
edge_points : list of int
Indices of the edge points.
"""
\# compute the kd-tree of the point cloud to enable fast neighbor search
kdtree = o3d.geometry.KDTreeFlann(pcd)
\# compute the normals of the point cloud
pcd.estimate_normals()
\# downsample the point cloud to speed up computation
downsampled_pcd = pcd.uniform_down_sample(self.uniform_downsample_factor)
edge_points_ind = []
for i in range(len(downsampled_pcd.points)):
\# find the neighbors of the point
_, point_indices, _ = kdtree.search_radius_vector_3d(downsampled_pcd.points[i], radius)
\# get the point cloud of the neighbors
neighbor_pcd = pcd.select_by_index(point_indices)
\# cluster the normals of the neighbors into two groups
orthogonal, _ = self.cluster_normals(neighbor_pcd)
\# if the normals could be clustered into two groups that are approximately orthogonal,
\# the point is considered an edge point
if orthogonal:
edge_points_ind.append(self.uniform_downsample_factor * i)
return edge_points_ind
def cluster_normals(self, pcd, tol_cluster=0.4):
Helper function for 'find_edge_points_normals'. Clusters the normals of a point cloud into
two groups. If the mean of one group is nearly parallel to the vertical direction, and the
mean of the other group is nearly orthogonal to the vertical direction, and there is
approximately an equal number of points in both groups, the point is considered an edge point.
Parameters

```
```

pcd : open3d PointCloud
Point cloud, where the normals should be clustered. Usually the neighbors of a point.
tol_cluster : float, optional
Tolerance of the inner product of the mean of the two clusters and the vertical direction.
If the inner product is 0, the rest is orthogonal to the up-vector. The default is 0.4.

```
Returns
orthogonal : bool
    The two clusters are approximately orthogonal to each other.
center_rest : ndarray of shape (3,)
    Mean of the cluster that is that is not the vertical direction.
    If the two clusters are not approximately orthogonal, None is returned.
"""
\# get the normals of the points in the point cloud
normals \(=\) np.asarray(pcd.normals)
\# tolerance for the length of the difference of the normals and the vertical direction
tol_parallel \(=0.3\)
\# all normals that are NOT approximately vertical (parallel to up_vector)
rest \(=\) normals \([\)
    np.logical_and (
        np.linalg.norm(normals - self.up_vector, axis=1) > tol_parallel,
        np.linalg.norm(normals + self.up_vector, axis=1) > tol_parallel,
        )
]
\# if there are too little normals of one kind, this is not an edge point
if \(4 * \operatorname{len}(r e s t)>\operatorname{len}(\) normals \()\) and \(4 * \operatorname{len}(r e s t)<3 * \operatorname{len}\) (normals):
    \# compute the mean and the norm of the rest
    center_rest = np.sum(rest, axis=0) / len(rest)
    norm_rest \(=\) np.linalg.norm(center_rest)
    \# tolerance for the length of the mean of the rest
    tol_norm \(=0.4\)
    \# if the norm is too short, the normals are too far away from each other -> return False
    if norm_rest > tol_norm:
        \# normalize the mean of the rest
        center_rest /= norm_rest
        \# compute the inner product of the mean of the rest and the vertical direction
        inner_product \(=\) center_rest @ self.up_vector
        \# if the inner product of the centers is close to zero,
        \# the clustered normals are nearly orthogonal
        if np.abs(inner_product) < tol_cluster:
            return True, center_rest
    return False, center_rest
return False, None
def differentiate_edges(self, pcd, edge1_P, edge2_P, lifting_vector=None):
    """
    Differentiate between a loading edge and the edge between loading platform and rear wall.
    Key difference: above the loading edge, there should be no points \(\rightarrow\) translate the edge a bit
        in the direction of the up_vector and check, which line hits less points -> loading edge
    Parameters
```

pcd : open3d PointCloud
Point cloud, where the edges should be differentiated.
edge1_P : ndarray of shape (2,3)
Two points that define the first edge.
edge2_P : ndarray of shape (2,3)
Two points that define the second edge.
lifting_vector : ndarray of shape (3,), optional
Vector in the direction of which the edges are translated. The default is None.
If None, the up_vector is used.

```

\section*{Returns}
```

loading_edge : ndarray of shape (2,3)

```
    Two points that define the loading edge.
other_edge : ndarray of shape \((2,3)\)
    Two points that define the other edge.
correct_order : bool
    True, if the loading edge is edge1_P, False if the loading edge is edge2_P.
"""
\# define the lifting vector
if lifting_vector is None:
    lifting_vector = self.up_vector
\# lift both edges by a certain factor
lifting_factor \(=0.3\)
lift = lifting_factor * lifting_vector
edge1_P_lifted \(=\) edge1_P + lift
edge2_P_lifted = edge2_P + lift
\# compute the number of points for both edges and both lifted edges
num_votes_lifted_1 = len(
        self.get_points_on_line(pcd, edge1_P_lifted, tol=0.1, cluster=False, between_points=True)
)
num_votes_lifted_2 = len(
        self.get_points_on_line(pcd, edge2_P_lifted, tol=0.1, cluster=False, between_points=True)
)
num_votes_1 = len(
        self.get_points_on_line(pcd, edge1_P, tol=0.1, cluster=False, between_points=True)
)
num_votes_2 = len(
        self.get_points_on_line(pcd, edge2_P, tol=0.1, cluster=False, between_points=True)
)
\# compute the relative votes of the lifted edges compared to the original edges
rel_votes1 = num_votes_lifted_1 / num_votes_1
rel_votes2 = num_votes_lifted_2 / num_votes_2
min_rel = 0.0001
if rel_votes1 < min_rel and rel_votes2 < min_rel:
        \# if both edges have little to no points above them, we take the longer edge
        if np.linalg.norm(edge1_P[1] - edge1_P[0]) > np.linalg.norm(edge2_P[1] - edge2_P[0]):
            loading_edge \(=\) edge1_P
            other_edge = edge2_P
            correct_order = True
        else:
            loading_edge = edge2_P
            other_edge = edge1_P
            correct_order = False
elif rel_votes1 < rel_votes2:
        \# edge1 is the loading edge
        loading_edge = edge1_P
```

    other_edge = edge2_P
    correct_order = True
    else:
\# edge2 is the loading edge
loading_edge = edge2_P
other_edge = edge1_P
correct_order = False
return loading_edge, other_edge, correct_order
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# Geometric Functions

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def line_ransac(self, pcd, tol=0.08, max_iter=100):
"""
Find a line in a point cloud by RANSAC. Typical usecase: The given point cloud consists of
previously detected edge points, and we want to find the dominant edge, i.e. the edge
containing the most points.
Parameters
pcd : open3d PointCloud
Point cloud, where the line should be detected.
Usually this point cloud consists of previously detected edge points.
tol : float, optional
Tolerance of the distance between a point and the line. The default is 0.08.
max_iter : int, optional
Maximum number of iterations of the RANSAC algorithm. The default is 100.
Returns
best_line : ndarray of shape (2,3)
Two points that define the line.
finished_ransac : bool
States if there are enough points in the point cloud to perform RANSAC.
"""
if len(pcd.points) > 1:
\# initialize the best line and the number of votes
best_votes = 0
best_line = None
for _ in range(max_iter):
\# sample the points and get the votes for this line
P1, P2, votes = self.line_ransac_votes(pcd, tol)
\# update the best line and the number of votes if necessary
if votes > best_votes:
best_votes = votes
best_line = np.array([P1, P2])
return best_line, True
else:
return None, False
def line_ransac_votes(self, pcd, tol, P=None):
"""
Helper function for 'line_ransac'. Counts the number of points that lie within some small
tolerance around a line. If P is None, two random points are sampled from pcd.
Parameters

```
```

pcd : open3d PointCloud
Point cloud, where the line should be detected.
tol : float
Tolerance of the distance between a point and the line.
P : list of ndarray of shape (2,3) or None, optional
If None, two random points are chosen and the line between them is used.
If not None, the line between the two points is used. The default is None.

```
Returns
-------
P1 : ndarray of shape (3,)
        First point of the line.
P2 : ndarray of shape (3,)
    Second point of the line.
votes : int
    Number of points that lie within the tolerance around the line.
" " " "
\# sample points if necessary
if \(P\) is None:
    ind1, ind2 = np.random.randint(len(pcd.points), size=2)
    P1 = pcd.points[ind1]
    P2 = pcd.points[ind2]
else:
    \(P 1=P[0]\)
    \(P 2=P[1]\)
\# compute the distance of each point to the line
vecs_to_points = pcd.points - P1
\# get the vector of the line
vec_line = P2 - P1
\# if the two points are too close to each other, we return zero votes
norm_line = np.linalg.norm(vec_line)
if norm_line < 1e-10:
    return P1, P2, 0
\# normalize the vector of the line
vec_line /= norm_line
\# project the vectors to the points onto the line
projected_length = vecs_to_points @ vec_line
\# to avoid numerical errors, we set negative values to zero
squared_dist = np.square(np.linalg.norm(vecs_to_points, axis=1)) - np.square(projected_length)
squared_dist[squared_dist < 0] = 0
dist_to_line = np.sqrt(squared_dist)
\# we count the number of points that lie within some small tolerance around the line
votes = np.count_nonzero(dist_to_line < tol)
return P1, P2, votes
def get_points_on_line(self, pcd, P, tol=0.05, cluster=True, between_points=False):
Returns the indices of the points of a point cloud that lie on a given line.
If cluster=True, the indices of the largest cluster of points on the line are returned.
If between_points=True, only the points between the two given points are returned.
Parameters
```

pcd : open3d PointCloud
Point cloud, where the points on the line should be found.
P : ndarray of shape (2,3)
Two points that define the line.
tol : float, optional
Tolerance of the distance between a point and the line. The default is 0.05.
cluster : bool, optional
If True, the indices of the largest cluster of points on the line are returned.
If False, all the indices of the points on the line are returned.
Only relevant, if between_points=False. The default is True.
between_points : bool, optional
If True, only the points between the two given points are returned.
If False, all the points on the line are returned. The default is False.
Returns
-------
points_on_line_ind : list of int
Indices of the points on the line.
"""

# get the two points that define the line

P1, P2 = P[0], P[1]

# get the distance of all points to the first point of the line

vecs_to_points = pcd.points - P1

# get the vector of the line and its length

vec_line = P2 - P1
original_len_line = np.linalg.norm(vec_line)

# normalize the vector of the line

vec_line /= original_len_line

# project the vectors to the points onto the vector of the line

projected_length = vecs_to_points @ vec_line
if between_points:
\# consider only points between P1 and P2
\# compute the distance of the points to the line with the Pythagorean theorem
squared_dist = np.square(np.linalg.norm(vecs_to_points, axis=1)) - np.square(
projected_length
)
\# avoid numerical errors and get the distance to the line
squared_dist[squared_dist < 0] = 0
dist_to_line = np.sqrt(squared_dist)
points_on_line_ind = []
for i in range(len(projected_length)):
if (
projected_length[i] > 0
and projected_length[i] < original_len_line
and dist_to_line[i] < tol
):
points_on_line_ind.append(i)
return points_on_line_ind
else:
\# consider all points
\# compute the distance of the points to the line with the Pythagorean theorem
squared_dist = np.square(np.linalg.norm(vecs_to_points, axis=1)) - np.square(
projected_length
)

```
```

        # avoid numerical errors and get the distance to the line
        squared_dist[squared_dist < 0] = 0
        dist_to_line = np.sqrt(squared_dist)
        # get the indices of the points that lie within the tolerance
        points_on_line_ind = np.where(dist_to_line < tol)[0]
        # return the indices of the largest cluster
        if cluster:
        return self.find_largest_cluster(pcd, points_on_line_ind)
        else:
            return points_on_line_ind
    def find_largest_cluster(self, pcd, pcd_ind=None, eps=0.5, min_points=10):
Find the largest cluster of a point cloud.
Parameters
pcd : open3d PointCloud
Point cloud, where the largest cluster should be found.
pcd_ind : list of int or None, optional
Indices of the points in pcd that we want to find a cluster in.
If None, we search for the largest cluster in the whole point cloud, e.g. if we want to
find the largest cluster of a line in a point cloud, pcd_ind gives the indices of the
points of the line in pcd. The default is None.
eps : float, optional
Epsilon for the DBSCAN algorithm. The default is 0.5.
min_points : int, optional
Minimum number of points for a cluster. The default is 10.
Returns
inlier_ind : list of int
Indices of the points of the largest cluster.
"""
if pcd_ind is None:
\# search for the largest cluster in the whole point cloud
pcd_ind = list(range(len(pcd.points)))
else:
\# or just in the points with the given indices
pcd = pcd.select_by_index(pcd_ind)

# cluster the pcd using dbscan

cluster_labels = np.array(pcd.cluster_dbscan(eps=eps, min_points=min_points))

# points labeled -1: noise

# get a list of all cluster labels

unique_labels = np.unique(cluster_labels)

# count for each cluster label the amount of points in that label

num_points_per_cluster = [len(np.where(cluster_labels == i)[0]) for i in unique_labels]

# If the two largest clusters have the same amount of points, we have to decide somehow

# between them. Generally, this is bad, because then there is no unique 'largest' cluster

# just take the first cluster -> [0] at the end

# (the first [0] gives us the first entry of the tuple (array,) that we get from np.where)

max_points_per_cluster = np.max(num_points_per_cluster)
max_label = np.where(num_points_per_cluster == max_points_per_cluster) [0] [0]

# get the label of the largest cluster

largest_cluster_label = int(unique_labels[max_label])

```
```


# get all the indices of pcd_ind that correspond to the points of the largest cluster

# in the larger point cloud that pcd is embedded in

inlier_ind = [pcd_ind[i] for i in list(np.where(cluster_labels == largest_cluster_label)[0])]
return inlier_ind
def find_orthogonal_line(self, edge_pcd, P_original, tol=0.08, max_iter=100, max_lines=5):
"""
For a given line, find a line that is approximately orthogonal to it within a point cloud.
The line is found by a version of RANSAC. At most max_lines lines are found.
The first line that is orthogonal is returned
If no orthogonal line is found, the original line is returned, together with False
Parameters
edge_pcd : open3d PointCloud
Point cloud, where the line should be detected.
Usually this point cloud consists of previously detected edge points.
P_orthogonal : ndarray of shape (2,3)
Two points that define the line that is orthogonal to the line we are looking for.
tol : float, optional
Tolerance of the distance between a point and the line. The default is 0.08.
max_iter : int, optional
Maximum number of iterations of the RANSAC algorithm for one line. The default is 100.
max_lines : int, optional
Maximum number of lines that are tried to find. The default is 5.
Return
best_line : ndarray of shape (2,3)
Two points that define the line.
orthogonal : bool
True, if an orthogonal line was found
"""

# initialize the best line and the number of votes

best_votes = 0
best_line = P_original

# define the original line

line_original = P_original[1] - P_original[0]
line_original /= np.linalg.norm(line_original)
for j in range(max_lines):
for _ in range(max_iter):
\# if the point cloud is too small, we cannot find a line
if len(edge_pcd.points) >= 2:
P1, P2, votes = self.line_ransac_votes(edge_pcd, tol)
else:
return P_original, False
\# update the best line and the number of votes if necessary
if votes > best_votes:
best_votes = votes
best_line = np.array([P1, P2])
\# compute the vector of the best line
vec_line = best_line[1] - best_line[0]
vec_line /= np.linalg.norm(vec_line)
\# if the line is orthogonal to the other line and to the vertical direction,
\# we have found the correct line

```
```

        if (
        np.abs(vec_line @ line_original) < self.tol_orthogonal
        and np.abs(vec_line @ self.up_vector) < self.tol_orthogonal
        ):
            return best_line, True
        else:
            # delete the points on this line from the edge point cloud and start again
            new_ind = self.get_points_on_line(edge_pcd, best_line, tol=0.05)
            edge_pcd = edge_pcd.select_by_index(new_ind, invert=True)
            best_votes = 0
            best_line = P_original
    # if no orthogonal line is found within the maximum number of iterations,
    # return the original line and False
    return P_original, False
    def find_endpoints(self, edge_pcd, max_iter=100):
"""
Find the endpoints of a line, i.e. the points on the line that are the furthest away from each
other. This is done by a variant of RANSAC. Usually a small number of iterations is enough,
since this does not have to be the best pair of points, a close approximation suffices.
If necessary, increase the number of iterations given by max_iter.
Parameters
edge_pcd : open3d PointCloud
Point cloud that contains the points on the edge.
This point cloud should already be the largest cluster of a found edge.
max_iter : int, optional
Maximum number of iterations. The default is 100.
Returns
best_P1 : ndarray of shape (3,)
First endpoint.
best_P2 : ndarray of shape (3,)
Second endpoint.
"""
\# initialize best distance and best points
best_dist = 0
best_P1 = None
best_P2 = None
\# find the best pair of points within max_iter iterations
for i in range(max_iter):
ind1, ind2 = np.random.randint(len(edge_pcd.points), size=2)
P1 = edge_pcd.points[ind1]
P2 = edge_pcd.points[ind2]
new_dist = np.linalg.norm(P2 - P1)
if new_dist > best_dist:
best_dist = new_dist
best_P1 = P1
best_P2 = P2
return best_P1, best_P2
"""
Additional functions, not shown here:
def intersect_line_plane(self, P_line, P_plane)
def intersect_two_planes(self, weights1, weights2)

```

\section*{\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\# \\ \# Alternative Approach}
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
def find_edge_points_centroids(self, pcd, radius=0.09, max_nn=40, tol=0.009):
    Find edge points of a point cloud by computing the centroid of the neighbors of each point.
    If the centroid is too far away from the point, the point is considered an edge point.
    Problem of this method: Finds also points at the border of the point cloud.
    Parameters
    pcd : open3d PointCloud
        Point cloud, where the edge points should be found.
    radius : float, optional
        Radius around a point, where the kd-tree looks for neighbors. The default is 0.09 .
    max_nn : int, optional
        Maximum number of neighbors to be found. The default is 40.
    tol: float, optional
        Tolerance of the distance between the centroid and the point. The default is 0.009.
        If the distance is larger than tol, the point is considered an edge point.
    Returns
    edge_points_ind : list of int
        Indices of the edge points.
    """
    \# compute the kd-tree of the point cloud to enable fast neighbor search
    kdtree \(=\) o3d.geometry.KDTreeFlann(pcd)
    \# downsample the point cloud to speed up computation
    downsampled_pcd = pcd.uniform_down_sample(self.uniform_downsample_factor)
    edge_points_ind = []
    for \(i\) in range(len(downsampled_pcd.points)):
        \# find the neighbors of the point
        _, point_indices, _ = kdtree.search_hybrid_vector_3d(
                query=downsampled_pcd.points[i], radius=radius, max_nn=max_nn
            )
            \# get the point cloud of the neighbors
            neighbor_pcd = pcd.select_by_index(point_indices)
            \# compute the centroid of the neighbors
            sum_of_points = np.sum(np.asarray(neighbor_pcd.points), axis=0)
            centroid = sum_of_points / len(neighbor_pcd.points)
            \# compute the distance between the centroid and the point
            dist_to_point = np.linalg.norm(centroid - downsampled_pcd.points[i])
            \# if the distance is larger than tol, the point is considered an edge point
            if dist_to_point > tol:
                edge_points_ind.append(self.uniform_downsample_factor * i)
    return edge_points_ind

\section*{B. Code Pose Estimation}
```

import gtsam
import numpy as np
import copy
import pickle

```
```

class Truck3D:

```
    def __init__(
        self,
        observation_noise_translation_default=0.17,
        observation_noise_rotation_default=0.14,
        construction_noise_translation_default=12,
        construction_noise_rotation_default=1.5,
        updating_factor_noise=0.94,
        updating_factor_weights=0.97,
        initial_weights_translation=0.8,
        initial_weights_rotation=0.8,
        align_every_n_steps=32,
        saving_old_factors_until_n_time_steps=50,
        **truck_configuration,
    ):
        " " "
    Constructor
    Parameters
    observation_noise_translation_default : float, optional
        Standard deviation of the noise for the translation of the observations.
        The default is 0.17.
    observation_noise_rotation_default : float, optional
        Standard deviation of the noise for the rotation of the observations. The default is 0.14.
    construction_noise_translation_default : float, optional
        Default standard deviation of the noise for the translation of the construction factors.
        Used for construction factors of parts that are in no special relation
        (dependency group or degree of freedom) with each other. The default is 12.
    construction_noise_rotation_default : float, optional
        Default standard deviation of the noise for the rotation of the construction factors.
        Used for construction factors of parts that are in no special relation
        (dependency group or degree of freedom) with each other. The default is 1.5.
    updating_factor_noise : float, optional
        Factor to update the construction noise after each time step.
        Update the default construction noise with this factor after each time step.
        The default is 0.94.
    updating_factor_weights : float, optional
        Factor to update the weights of the factors after each time step.
        The default is 0.97.
    initial_weights_translation : float, optional
        Initial weight for updating the translation of parts of the truck after a new observation.
        The default is 0.8.
    initial_weights_rotation : float, optional
        Initial weight for updating the rotation of parts of the truck after a new observation.
        The default is 0.8 .
    align_every_n_steps : int, optional
        How often dependent parts of the model are aligned.

If 0 , dependent parts are never aligned.
If 1, dependent parts are aligned after each time step. The default is 32.
saving_old_factors_until_n_time_steps : int, optional
Number of time steps, old observation factors should stay in the graph.
The default is 50.
**truck_configuration : dict
Dictionary with the configuration of the truck. The keys of the dictionary are some
features of the truck (e.g. width of the loading platform). Not all features
have to be given (for the features that are not given, default values are used).

\section*{Attributes}
graph : gtsam.NonlinearFactorGraph
Current factor graph.
updating_factor_noise : float in (0, 1]
Factor to update the construction noise after each time step.
The construction noise is updated with the function update_construction_noise().
updating_factor_weights : float in (0, 1]
Factor to update the weights for model updates after each time step.
The weights are updated with the function update_weights().
lower_bound_construction_noise : float >= 0
Lower bound for the construction noise in the updating function.
lower_bound_weights : float in [0, 1]
Lower bound for the weights in the updating function.
weight_rotation : float in (lower_bound_weights, 1]
Current weight for the updates of the rotation.
weight_translation : float in (lower_bound_weights, 1]
Current weight for the updates of the translation.
observation_noise_rotation_default : float >= 0
Standard deviation of the noise for the rotation of the observations.
observation_noise_translation_default : float >= 0
Standard deviation of the noise for the translation of the observations.
construction_noise_rotation_default : float
Default standard deviation of the noise for the rotation of the construction factors. Used for construction factors of parts that are in no special relation to each other.
construction_noise_rotation_dependent : float
Standard deviation of the noise for the rotation of the construction factors between dependent parts.
construction_noise_translation_default : float
Default standard deviation of the noise for the translation of the construction factors.
Used for construction factors of parts that are in no special relation to each other.
construction_noise_translation_dependent : float
Standard deviation of the noise for the translation of the construction factors between dependent parts.
align_every_n_steps : int >= 0
How often dependent parts of the model are aligned.
If 0 , dependent parts are never aligned.
If 1, dependent parts are aligned after each time step.
factors_to_reuse : list of gtsam.NonlinearFactorGraph
List of the observation factors that are saved to reuse them after each time step
in the recreation of the factor graph. Old observations stay the same each time
the factor graph is recreated and should be kept in the factor graph for some time.
saving_old_factors_until_n_time_steps : int >0
Number of time steps, old observation factors should stay in the graph.
time_step_factors_to_reuse : list of int List of the corresponding time step of the factors to reuse.
Used to check if observations are too old and therefore deleted from the list.
symbols_sensor : list of gtsam.Symbol
List of the symbols for the sensor in each time step.
sensor_trajectory : list of gtsam.Pose3
List of the estimated poses of the sensor in the world frame.
current_num_time_steps : int

Number of time steps that have already been processed.
g0, GO, w1, w2, w3, W1, W2, W3, l1, l2, L1, L2 : gtsam.Symbol
Symbols for the variables.
g0... origin of the world/global frame
GO ... not any origin, but to stay consistent with the naming convention
w1, w2, w3 ... left wheels (one front wheel (w1) and two rear wheels (w2, w3))
W1, W2, W3 ... right wheels (one front wheel (W1) and two rear wheels (W2, W3))
l1, l2... left lights (one front light (l1) and one rear light (l2))
L1, L2 .. right lights (one front light (L1) and one rear light (L2))
symbols_parts : list of gtsam.Symbol
List of all symbols for the variables.
dependency_groups : dict
Dependency groups contain parts of the truck that have a fixed spatial relation
to each other, e.g. all wheels have the same z-coordinate in the world frame.
Therefore, if the \(z\)-coordinate of one wheel is changed, the z-coordinates of all other
wheels have to change as well. Given as a dictionary with the axes as keys,
where the values are dictionarys with the dependency groups as values.
groups_with_degrees_of_freedom : dict
Some parts have certain degrees of freedom, e.g. the front wheels can be rotated around the z-axis (but both front wheels are rotated the same amount). Given as a dictionary with the axes as keys. The values are again dictionarys containing groups of symbols that are in a fixed relation to each other. A group is given as a dictionary containing "symbols" (list of symbols) and "std" (additional
standard deviation of the noise model). If a single element has a degree of freedom,
it can be added here as well (e.g. a wheel can be rotated around the \(y\)-axis).
symbols_info : dict
Dictionary with information about each symbol defining the model of the truck implicitly.
For each symbol, the following information is contained:
num_observed: integer, indicating how often the symbol has been observed.
pose_in_world_frame: dictionary with keys " \(x\) ", " \(y\) ", " \(z\) ", "rotation_matrix".
dependency_groups: dictionary with keys "x", "y", "z", "roll", "pitch", "yaw" and the symbols in their respective group as values.
groups_with_degrees_of_freedom: dictionary with only the necessary keys of " \(x\) ", " \(y\) ",
"z", "roll", "pitch", and "yaw" and the symbols in their respective group as values.
std_degrees_of_freedom: dictionary with the additional standard deviation for
the respective groups with degrees of freedom.
construction_noise_dictionary : dict
Dictionary with the standard deviations of the noise models for the construction factors.
This information is saved such that it does not have to be computed in each time step.
length_loading_platform : float
Length of the loading platform.
true_truck_configuration : dict
Dictionary that contains the ground truth of the truck configuration. Used for testing. """
self.graph \(=\) gtsam.NonlinearFactorGraph()
\# define input parameters for noise, weights, and updating factors as attributes
self.updating_factor_noise = updating_factor_noise
self.updating_factor_weights = updating_factor_weights
self.lower_bound_construction_noise \(=0.1\)
self.lower_bound_weights = 0.1
self.weight_rotation = initial_weights_rotation
self.weight_translation = initial_weights_translation
self.observation_noise_rotation_default = observation_noise_rotation_default
self.observation_noise_translation_default = observation_noise_translation_default
self.construction_noise_rotation_default = construction_noise_rotation_default
self.construction_noise_rotation_dependent \(=0\)
self.construction_noise_translation_default = construction_noise_translation_default
self.construction_noise_translation_dependent \(=0\)
\# define how often dependent parts of the model are aligned
self.align_every_n_steps = align_every_n_steps
```


# define empty lists for recreating the factor graph

self.factors_to_reuse = []
self.saving_old_factors_until_n_time_steps = saving_old_factors_until_n_time_steps
self.time_step_factors_to_reuse = []

# define empty lists for the symbols and the estimated poses of the sensor

self.symbols_sensor = []
self.sensor_trajectory = []

# define a counter for the number of time steps

self.current_num_time_steps = 0

# define all available symbols for the variables

(
self.g0,
self.GO,
self.w1,
self.w2
self.w3,
self.W1,
self.W2,
self.W3,
self.l1,
self.l2,
self.L1,
self.L2,
) = self.define_symbols()
self.symbols_parts = [
self.g0,
self.GO,
self.w1,
self.w2,
self.w3,
self.W1,
self.W2,
self.W3
self.l1,
self.l2
self.L1,
self.L2,
]

# dictionary with symbols that are spatially dependent on each other in some way (i.e. axis)

self.dependency_groups = {
"x": {
: [self.g0, self.GO],
[self.w1, self.W1],
[self.w2, self.W2],
[self.w3, self.W3],
[self.l1, self.L1],
},
"y": {
0: [self.w1, self.w3],
1: [self.W1, self.W3],
},
"z": {
0: [self.W1, self.W2, self.w3, self.W1, self.W2, self.W3],
[ [self.l1, self.L1]
: [self.l2, self.L2],
},
}

# dictionary with symbols that have some degrees of freedom in their relation to other symbols

```
```


# and the parts that are in a fixed relation to them

# the additional uncertainty is given as the standard deviation of the noise model ("std")

self.groups_with_degrees_of_freedom = {
"yaw": {
\# the yaw of the front wheels might change depending on the steering angle
0: {
"symbols": [self.w1, self.W1],
"std": 1.5 * np.pi,
"rotate_second_matrix": np.pi,
},
},
"z": {
\# the height of the loading edge might change depending on the load
0: {"symbols": [self.gO, self.GO], "std": 2},
},
}

# dictionary with information about the variables

self.symbols_info = {}
self.build_symbols_info()

# get the construction noise model depending on the degrees of freedom and dependent parts

self.construction_noise_dictionary = {}
self.build_construction_noise_dictionary()

# define the approximate truck configuration

self.length_loading_platform = 5.87
self.define_estimated_truck_configuration(truck_configuration)

# define the ground truth configuration of the truck

self.true_truck_configuration = {}
self.define_ground_truth_truck_configuration()
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#

# Definition of the truck

\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
def build_symbols_info(self):
Build the symbols_info dictionary at the initialization of the class.
It contains for each symbol the number of times this symbol has been observed,
the poses of the parts in the world frame,
the dependency groups the symbol is involved in,
and the groups with degrees of freedom the symbol is involved in
with the respective standard deviation.
"""
self.symbols_info = {
symbol: {
"num_observed": 0,
"pose_in_world_frame": {
"x": 0,
"y": 0,
"z": 0,
"rotation_matrix": gtsam.Rot3(),
},
"dependency_groups": {
"x": [],
"y": [],
"z": [],
"roll": [],
"pitch": [],
"yaw": [],
},

```
```

            "groups_with_degrees_of_freedom": {},
            "std_degrees_of_freedom": {},
        }
        for symbol in self.symbols_parts
    }

# adapt the dependency groups and groups with degrees of freedom for each symbol

for symbol in self.symbols_parts:
for axis in self.dependency_groups:
for group in self.dependency_groups[axis]:
\# if the symbol is in this specific group, we add all other symbols in this group
\# to the dependency groups of the symbol
if symbol in self.dependency_groups[axis][group]:
for other_symbol in self.dependency_groups[axis] [group]:
if other_symbol != symbol:
self.symbols_info[symbol]["dependency_groups"] [axis].append(
other_symbol
)

```
    for axis in self.groups_with_degrees_of_freedom:
    for group in self.groups_with_degrees_of_freedom[axis]:
            if symbol in self.groups_with_degrees_of_freedom[axis] [group] ["symbols"]
                \# if the symbol is in this specific group, we add all other symbols in this
                \# group to the dependency groups of the symbol, and we add the standard
                \# deviation of the degrees of freedom to the symbols_info dictionary
                \# gather all other symbols in this group
                other_symbols = []
                for other_symbol in self.groups_with_degrees_of_freedom[axis] [group] [
                "symbols"
                ]:
                if other_symbol != symbol:
                    other_symbols.append(other_symbol)
                self.symbols_info[symbol]["groups_with_degrees_of_freedom"][
                axis
                ] = other_symbols
                self.symbols_info[symbol]["std_degrees_of_freedom"][
                axis
                ] = self.groups_with_degrees_of_freedom[axis] [group] ["std"]
def define_estimated_truck_configuration(self, truck_configuration=\{\}):
    """
    Define the estimated configuration of the truck by defining the poses of all parts
    in the world frame in the symbols_info dictionary.
    Calls the function define_symbols_info_for_estimated_truck_configuration().
    Parameters
    truck_configuration : dict, optional
        Dictionary with the configuration of the truck. The keys of the dictionary are some
        features of the truck (e.g. width of the loading platform). Not all features have to be
        given (for the features that are not given, default values are used). The default is \{\}.
        " " "
    \# we have to check whether the length of the loading platform is given
    if "length_loading_platform" not in truck_configuration:
        self.length_loading_platform \(=5.87\)
    else:
        self.length_loading_platform = truck_configuration["length_loading_platform"]
    \# define the poses of all parts in the world frame in the symbols_info dictionary
    \# pass the truck configuration with ''**'' to unpack the dictionary
    self.define_symbols_info_for_estimated_truck_configuration(**truck_configuration)
def define_symbols_info_for_estimated_truck_configuration(
self,
rotation_matrix_right_loading_edge=None,
rotation_matrix_right_wheels=None,
rotation_matrix_front_lights=None,
rotation_matrix_rear_lights=None,
width_loading_platform=2.47,
height_loading_platform_above_wheels=0.70,
distance_g0_w1=0.79,
distance_w1_w2=3.84,
distance_w2_w3=1.35,
height_loading_platform_above_front_lights=0.38,
height_loading_platform_above_rear_lights=0.46,
distance_lights_in_front_of_loading_platform=2.19,
distance_lights_behind_loading_platform=0.79,
distance_front_lights=1.73,
distance_rear_lights=1.96,
):
" " "
Define the poses of all parts in the world frame in the symbols_info dictionary.
This function is a helper function for define_estimated_truck_configuration().
Parameters
rotation_matrix_right_loading_edge : ndarray, optional
Rotation matrix for the right loading edge. The default is None.
rotation_matrix_right_wheels : ndarray, optional
Rotation matrix for the right wheels. The default is None.
rotation_matrix_front_lights : ndarray, optional
Rotation matrix for the front lights. The default is None.
rotation_matrix_rear_lights : ndarray, optional
Rotation matrix for the rear lights. The default is None.
width_loading_platform : float, optional
Width of the loading platform. The default is 2.47.
height_loading_platform_above_wheels : float, optional
Height of the loading platform above the wheels. The default is 0.70.
distance_gO_w1: float, optional
Distance between the origin of the world frame and the left front wheel.
The default is 0.79.
distance_w1_w2 : float, optional
Distance between the left front wheel and the left rear wheel. The default is 3.84.
distance_w2_w3 : float, optional
Distance between the left rear wheel and the right rear wheel. The default is 1.35.
height_loading_platform_above_front_lights : float, optional
Height of the loading platform above the front lights. The default is 0.38.
height_loading_platform_above_rear_lights : float, optional
Height of the loading platform above the rear lights. The default is 0.46.
distance_lights_in_front_of_loading_platform : float, optional
Distance between the front lights and the loading platform. The default is 2.19.
distance_lights_behind_loading_platform : float, optional
Distance between the rear lights and the loading platform. The default is 0.79.
distance_front_lights : float, optional
Distance between the two front lights. The default is 1.73.
distance_rear_lights : float, optional
Distance between the two rear lights. The default is 1.96.
"""
\# if no rotation matrices are given, we use the default matrices
if rotation_matrix_right_loading_edge is None:
rotation_matrix_right_loading_edge = self.get_rotation_matrix(-np.pi / 2, axis="z")
if rotation_matrix_right_wheels is None:
rotation_matrix_right_wheels = self.get_rotation_matrix(np.pi, axis="z")
if rotation_matrix_front_lights is None:
```

    rotation_matrix_front_lights = self.get_rotation_matrix(np.pi / 2, axis="z")
    if rotation_matrix_rear_lights is None:
rotation_matrix_rear_lights = self.get_rotation_matrix(-np.pi / 2, axis="z")
y_middle = width_loading_platform / 2

# define the poses of all parts in the world frame in the symbols_info dictionary

self.symbols_info[self.g0]["pose_in_world_frame"] = {
"x": 0,
"y": 0,
"z": 0,
"rotation_matrix": gtsam.Rot3(),
}
self.symbols_info[self.GO]["pose_in_world_frame"] = {
"x": 0,
"y": width_loading_platform,
"z": 0,
"rotation_matrix": gtsam.Rot3(rotation_matrix_right_loading_edge),
}
self.symbols_info[self.w1]["pose_in_world_frame"] = {
"x": -distance_g0_w1,
"y": 0,
"z": -height_loading_platform_above_wheels,
"rotation_matrix": gtsam.Rot3(),
}
self.symbols_info[self.W1]["pose_in_world_frame"] = {
"x": -distance_g0_w1,
"y": width_loading_platform,
"z": -height_loading_platform_above_wheels,
"rotation_matrix": gtsam.Rot3(rotation_matrix_right_wheels),
}
self.symbols_info[self.w2]["pose_in_world_frame"] = {
"x": -distance_g0_w1 + distance_w1_w2,
"y": 0,
"z": -height_loading_platform_above_wheels,
"rotation_matrix": gtsam.Rot3(),
}
self.symbols_info[self.W2]["pose_in_world_frame"] = {
"x": -distance_g0_w1 + distance_w1_w2,
"y": width_loading_platform,
"z": -height_loading_platform_above_wheels,
"rotation_matrix": gtsam.Rot3(rotation_matrix_right_wheels),
}
self.symbols_info[self.w3]["pose_in_world_frame"] = {
"x": -distance_g0_w1 + distance_w1_w2 + distance_w2_w3,
"y": 0,
"z": -height_loading_platform_above_wheels,
"rotation_matrix": gtsam.Rot3(),
}
self.symbols_info[self.W3]["pose_in_world_frame"] = {
"x": -distance_g0_w1 + distance_w1_w2 + distance_w2_w3,
"y": width_loading_platform,
"z": -height_loading_platform_above_wheels,
"rotation_matrix": gtsam.Rot3(rotation_matrix_right_wheels),
}
self.symbols_info[self.l1]["pose_in_world_frame"] = {
"x": -distance_lights_in_front_of_loading_platform,
"y": y_middle - distance_front_lights / 2,
"z": -height_loading_platform_above_front_lights,
"rotation_matrix": gtsam.Rot3(rotation_matrix_front_lights),
}
self.symbols_info[self.L1]["pose_in_world_frame"] = {
"x": -distance_lights_in_front_of_loading_platform,

```
```

        "y": y_middle + distance_front_lights / 2,
        "z": -height_loading_platform_above_front_lights,
        "rotation_matrix": gtsam.Rot3(rotation_matrix_front_lights),
    }
    self.symbols_info[self.l2]["pose_in_world_frame"] = {
    "x": self.length_loading_platform + distance_lights_behind_loading_platform,
    "y": y_middle - distance_rear_lights / 2,
    "z": -height_loading_platform_above_rear_lights,
    "rotation_matrix": gtsam.Rot3(rotation_matrix_rear_lights),
    }
    self.symbols_info[self.L2]["pose_in_world_frame"] = {
    "x": self.length_loading_platform + distance_lights_behind_loading_platform,
    "y": y_middle + distance_rear_lights / 2,
    "z": -height_loading_platform_above_rear_lights,
    "rotation_matrix": gtsam.Rot3(rotation_matrix_rear_lights),
    }
    """
Additional functions, not shown here:
def define_symbols(self)
def get_sensor_symbol(self, time_step)
def define_ground_truth_truck_configuration(self)
def symbols_info_to_pose_in_world_frame(self, symbol, truck_configuration, true_or_estimated)
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#

# Update of the truck configuration

\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#
def update_truck_configuration(self, result):
"""
Update the poses of the parts of the truck in the world frame according to the weights in the
symbols_info dictionary. The model of the truck is updated after each time step, when new
observations lead to new optimization results and a new estimation of the truck configuration.
If we want to align the dependent parts explicitly, we can set align_explicitly to True.
Parameters
result : gtsam.Values
Result of the optimization containing the new poses of the parts in the world frame.
"""
for symbol in self.symbols_info.keys():
\# for all variables (parts of the truck and recent sensors): get the new pose from the
\# optimization result and update the pose in the symbols_info dictionary
new_pose = result.atPose3(symbol)
self.update_pose_in_world_frame(symbol, new_pose)
\# update the length of the loading platform, which is dependent on the poses of some parts
self.update_dependent_parameters()
\# align the dependent parts explicitly if desired
align_explicitly = False
if self.align_every_n_steps > 0 and self.current_num_time_steps > 0:
if self.current_num_time_steps % self.align_every_n_steps == 0:
align_explicitly = True
if align_explicitly:
self.align_dependent_parts()
\# transform the whole model such that the origin of the world frame
\# is at the beginning of the left loading edge
trafo_g0_to_origin = self.symbols_info_to_pose_in_world_frame(self.g0).between(gtsam.Pose3())

```
```

    for symbol in self.symbols_info.keys():
    new_pose = self.symbols_info_to_pose_in_world_frame(symbol).compose(trafo_g0_to_origin)
    self.symbols_info[symbol]["pose_in_world_frame"] = {
            "x": new_pose.translation() [0],
            "y": new_pose.translation() [1],
            "z": new_pose.translation() [2],
            "rotation_matrix": new_pose.rotation(),
        }
    def update_pose_in_world_frame(self, symbol, new_pose):
"""
Update the pose of a part in the world frame according to the updating weights.
Parameters
symbol : gtsam.Symbol
Symbol of the part.
new_pose : gtsam.Pose3
New estimated pose.
"""
old_pose = self.symbols_info_to_pose_in_world_frame(symbol)
\# update the rotation and translation according to the weights
updated_rotation = self.update_rotation(old_pose, new_pose, self.weight_rotation)
updated_translation = self.update_translation(old_pose, new_pose, self.weight_translation)
\# update the pose in the symbols_info dictionary
self.symbols_info[symbol]["pose_in_world_frame"] = {
"x": updated_translation[0],
"y": updated_translation[1],
"z": updated_translation[2],
"rotation_matrix": updated_rotation,
}
def update_rotation(self, old_pose, new_pose, weight):
"""
Update the rotation of a pose according to some weight.
Parameters
----------
old_pose : gtsam.Pose3
Old pose.
new_pose : gtsam.Pose3
New pose.
weight : float
Weight for the update.
Returns
-------
gtsam.Rot3
Updated rotation.
"""
rotation_matrix_new = new_pose.rotation().matrix()
rotation_matrix_old = old_pose.rotation().matrix()
\# compute the naive weighted average
average_rotation_matrix = weight * rotation_matrix_new + (1 - weight) * rotation_matrix_old
\# project this matrix onto SO(3)
\# compute the singular value decomposition of the matrix
U, _, V_t = np.linalg.svd(average_rotation_matrix, full_matrices=False)
S = np.diag([1, 1, np.sign(np.linalg.det(U) * np.linalg.det(V_t))])

```
average_rotation_matrix \(=\mathrm{U}\) @ S @ V_t
return gtsam. Rot3(average_rotation_matrix)
def update_translation(self, old_pose, new_pose, weight):
"""
Update the translation of a pose according to some weight.
Parameters
old_pose : gtsam.Pose3 Old pose.
new_pose : gtsam.Pose3 New pose.
weight : float Weight of the new pose.

Returns
gtsam.Point3 Updated translation.
" " "
return weight * new_pose.translation() + (1 - weight) * old_pose.translation()
def update_dependent_parameters(self):
""""
After changing the configuration of the truck by changing the poses of some parts
in the world frame, we have to update the parameters dependent on these poses,
i.e. the length of the loading platform. The new length is computed as the weighted average
of certain distances. Parts that have been observed more often get more weight.
" " "
weight_g0 \(=\max \left(s e l f . s y m b o l s \_i n f o[s e l f . g 0]\left[" n u m \_o b s e r v e d "\right], ~ 1\right) ~\)
weight_G0 \(=\max \left(s e l f . s y m b o l s \_i n f o[s e l f . G 0]\left[" n u m \_o b s e r v e d "\right], ~ 1\right) ~\)
weight_12 = max (self.symbols_info[self.l2] ["num_observed"], 1)
weight_L2 \(=\) max (self.symbols_info[self.L2]["num_observed"], 1)
x_front_end \(=\) (
weight_g0 * self.symbols_info[self.g0]["pose_in_world_frame"] ["x"]
+ weight_GO * self.symbols_info[self.GO]["pose_in_world_frame"] ["x"]
) / (weight_g0 + weight_GO)
x_rear_end = (
weight_l2 * self.symbols_info[self.12]["pose_in_world_frame"] ["x"]
+ weight_L2 * self.symbols_info[self.L2]["pose_in_world_frame"]["x"]
) / (weight_12 + weight_L2)
distance_lights_behind_loading_platform = 1.04
self.length_loading_platform = (
abs(x_rear_end - x_front_end) - distance_lights_behind_loading_platform
)
def align_dependent_parts(self):
"""
If the position of a part is updated, the positions of the dependent parts have to be updated as well, e.g. if the \(x\)-coordinate of the wheel w2 is updated, the \(x\)-coordinate of the wheel W2 has to be updated as well. This is done according to the align_every_n_steps parameter.
"""
\# we iterate over all axes ...
for axis in self.dependency_groups.keys():
\# ... and all groups to calculate the mean of the updated symbols in the current group
for group_list in self.dependency_groups[axis].values():
\# average for translation axes
if axis in ["x", "y", "z"]:
average_translation = self.get_average_translation(group_list, axis)
```

                    for symbol in group_list:
                        self.symbols_info[symbol] ["pose_in_world_frame"] [axis] = average_translation
        # average for rotation axes
        else:
            average_rotation_matrix, _ = self.get_average_rotation(group_list)
            for symbol in group_list:
            self.symbols_info[symbol]["pose_in_world_frame"] [
                    "rotation_matrix"
            ] = gtsam.Rot3(average_rotation_matrix)
    # and do the same for the groups with degrees of freedom
    for axis in self.groups_with_degrees_of_freedom.keys():
        for group in self.groups_with_degrees_of_freedom[axis].values():
        group_list = group["symbols"]
        if axis in ["x", "y", "z"]:
            average_translation = self.get_average_translation(group_list, axis)
            for symbol in group_list:
                    self.symbols_info[symbol] ["pose_in_world_frame"] [axis] = average_translation
        else:
            # we have to check if the rotation should be the same or rotated by a fixed angle
            average_rotation_matrix1, average_rotation_matrix2 = self.get_average_rotation(
            group_list, axis, rotate_second_matrix=group["rotate_second_matrix"]
            )
            self.symbols_info[group_list[0]]["pose_in_world_frame"][
                "rotation_matrix"
            ] = gtsam.Rot3(average_rotation_matrix1)
            self.symbols_info[group_list[1]]["pose_in_world_frame"][
                "rotation_matrix"
            ] = gtsam.Rot3(average_rotation_matrix2)
    def get_average_translation(self, list_of_symbols, axis):
"""
Get the average translation of a list of symbols in a certain direction.
The poses are weighted according to how often they have been observed.
Parameters
list_of_symbols : list of gtsam.Symbol
The list of symbols, we want to find the average of.
axis : str
The axis for which we want to find the average. One of ["x", "y", "z"].
Returns
-------
weighted_pose_average : float
The average translation in the given direction.
"""
sum_pose, sum_observed = 0, 0
for symbol in list_of_symbols:
obs = max(self.symbols_info[symbol]["num_observed"], 1)
sum_pose += self.symbols_info[symbol]["pose_in_world_frame"][axis] * obs
sum_observed += obs
weighted_pose_average = sum_pose / sum_observed
return weighted_pose_average
def get_average_rotation(self, list_of_symbols, axis=None, rotate_second_matrix=False):

```

Get the average rotation of a list of rotation matrices. If the rotations are dependent on each other, but rotated by a fixed angle, one matrix is rotated by this angle, then they are averaged, and then this matrix is rotated back.

Parameters
list_of_symbols : list of gtsam.Symbol
The list of symbols, we want to find the average of.
axis : str
The axis around which we want to find the average. One of ["roll", "pitch", "yaw"]. rotate_second_matrix : bool or angle, optional

If not False, this is the angle around which the second rotation matrix is rotated.
```

Returns
average_rotation_matrix : ndarray
The average rotation around the given axis.
""
if rotate_second_matrix is not False:
rot_matrix1 = self.symbols_info[list_of_symbols[0]]["pose_in_world_frame"][
'rotation_matrix"
].matrix()
rot_matrix2 = self.symbols_info[list_of_symbols[1]]["pose_in_world_frame"][
"rotation_matrix"
].matrix()
angle = rotate_second_matrix
helper_rot_matrix = self.get_rotation_matrix(angle, axis)
rot_matrix2 = helper_rot_matrix @ rot_matrix2
\# compute the naive average
average_rotation_matrix = (rot_matrix1 + rot_matrix2) / 2
else
list_of_matrices = []
for symbol in list_of_symbols:
list_of_matrices.append(
self.symbols_info[symbol]["pose_in_world_frame"]["rotation_matrix"].matrix()
)
\# compute the naive average
average_rotation_matrix = np.sum(np.array(list_of_matrices), axis=0) / len(
list_of_matrices
)

# project this matrix onto SO(3) and compute the singular value decomposition

U, _, V_t = np.linalg.svd(average_rotation_matrix, full_matrices=False)
S = np.diag([1, 1, np.sign(np.linalg.det(U) * np.linalg.det(V_t))])
average_rotation_matrix = U @ S @ V_t
if rotate_second_matrix:
return average_rotation_matrix, helper_rot_matrix.T @ average_rotation_matrix
else
return average_rotation_matrix, None
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#

# Pose estimation

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def estimate_sensor_pose(self, observations, observations_info=None, random_guess=False):
"""
Estimate the pose of the sensor in the world frame that fits the observations best.

```

Parameters
observations : dict
Dictionary with the symbols of the observed parts as keys and the observations as values. observations_info : dict, optional

Dictionary with the symbols of the observed parts as keys and the information about the
observations as values. Such information is an axis with certain degree of freedom due to
incomplete observations, e.g. we only observe a part of the loading edge
but not its beginning. The default is None.
random_guess : bool, optional
If True or the sensor trajectory is empty, the initial guess for the optimization of the sensor pose is a random pose.
Else, the initial guess for the optimization is the estimated pose of the sensor in the previous time step.
The default is False.

Returns
best_pose_new_sensor : gtsam.Pose3
Estimated pose of the sensor in the world frame.
best_error : float
Error of the optimization. Note: This is not the error of the estimated sensor pose
to the ground truth, but the error of the optimization.
result : gtsam.Values
Result of the optimization.
"""
\# define initial guess for the optimization
if random_guess or self.current_num_time_steps == 0:
initial_guess_sensor_pose = gtsam.Pose3(
gtsam.Rot3(), gtsam.Point3(np.random.rand(3) * 10)
)
else:
initial_guess_sensor_pose = self.get_current_sensor_pose()
\# insert the observations as factors in the factor graph
self.observe(observations, observations_info)
\# get the initial values for all parts and previous sensor poses
initial = self.get_initial_values()
\# estimate the pose of the sensor
best_pose_new_sensor, best_error, result = self.optimize_sensor_pose(
initial, initial_guess_sensor_pose
)
\# add the estimated pose to the sensor trajectory and the symbols_info dictionary
self.add_pose_to_trajectory(best_pose_new_sensor)
return best_pose_new_sensor, best_error, result
def observe(self, observations, observations_info=None):
" " " "
First, we recreate the factor graph with the new estimated truck configuration.
Then, we add a new sensor to the factor graph.
Finally, add the observations as factors to the factor graph.

Parameters
observations : dict
Dictionary with the symbols of the observed parts as keys and the observations as values. observations_info : dict, optional

Dictionary with the symbols of the observed parts as keys and the information about the observations as values. Such information is an axis with certain degree of freedom due to
incomplete observations, e.g. we only observe a part of the loading edge but not its beginning. The default is None.

\section*{"""}
\# create the factor graph from scratch (this substitutes the old factor graph)
self.create_factor_graph()
\# add the old observation factors
indices_to_delete = []
for \(i\) in range(len(self.factors_to_reuse)): if (
self.time_step_factors_to_reuse[i]
< self.current_num_time_steps - self.saving_old_factors_until_n_time_steps ):
indices_to_delete.append(i)
\# the list is monotonically increasing, so we can break here else:
break
\# delete the old factors from the list of factors to reuse
for i in reversed(indices_to_delete):
del self.factors_to_reuse[i]
del self.time_step_factors_to_reuse[i]
\# add the old factors to the new factor graph
for factor in self.factors_to_reuse:
self.graph.add(factor)
\# we also have to remove the old sensor symbol from the dictionary of all variables
if self.current_num_time_steps - self.saving_old_factors_until_n_time_steps > 0 :
self.symbols_info.pop (
self.get_sensor_symbol(
self.current_num_time_steps - self.saving_old_factors_until_n_time_steps - 1 )
)
\# add a new sensor and the observations as factors to the factor graph
self.add_new_sensor(observations, observations_info)
def add_new_sensor(self, observations, observations_info):
"""
Add a new sensor and the observations to the factor graph of the current time step.

Parameters
observations : dict
Dictionary with the symbols of the observed parts as keys and the observations as values. observations_info : dict, optional

Dictionary with the symbols of the observed parts as keys and the information about the observations as values. Such information is an axis with certain degree of freedom due to incomplete observations, e.g. we only observe a part of the loading edge but not its beginning. The default is None.
"""
\# get the symbol for the sensor in the next time step and add it to the list of sensor symbols next_time_step = self.current_num_time_steps
symbol_new_sensor = self.get_sensor_symbol (next_time_step)
self.symbols_sensor.append (symbol_new_sensor)
\# add the observations as factors to the factor graph
self.observations_to_factors(symbol_new_sensor, observations, observations_info)
def observations_to_factors(self, sensor_symbol, observations, observations_info): """

Convert the observations to factors between the sensor and the parts and insert them into the factor graph. Used in the function add_new_sensor(). The observations_info dictionary gives additional information about the observations, i.e. uncertainty in certain directions,
e.g. observations_info \(=\{w 1:\{" x ": 2.4\), "roll": 0.6*np.pi\}\}
... if the observation of w1 is unsure in \(x\)-direction and roll rotation.

\section*{Parameters}
sensor_symbol : gtsam.Symbol
Symbol of the sensor in the current time step.
observations : dict
Dictionary with the symbols of the observed parts as keys and the observations as values.
observations_info : dict
Dictionary with the symbols of the observed parts as keys and the information about the observations as values. Such information is an axis with certain degree of freedom due to incomplete observations, e.g. we only observe a part of the loading edge but not its beginning. The default is None.
" " "
for symbol in observations:
\# define the observation noise model
observation_noise_model = \{
"roll": self.observation_noise_rotation_default,
"pitch": self.observation_noise_rotation_default,
"yaw": self.observation_noise_rotation_default,
"x": self.observation_noise_translation_default,
"y": self.observation_noise_translation_default,
"z": self.observation_noise_translation_default,
\(\}\)
\# adapt the observation noise model according to observations_info
if observations_info is not None:
if symbol in observations_info.keys():
for axis in observations_info[symbol].keys():
observation_noise_model[axis] = observations_info[symbol] [axis]
\# define the observation noise model
observation_noise_model = self.get_noise_model(**observation_noise_model)
\# define the factor and add it to the graph
factor \(=\) gtsam. BetweenFactorPose3(
sensor_symbol, symbol, observations[symbol], observation_noise_model
)
self.graph.add(factor)
\# add the factor to the list of factors that should be reused in the next time step
self.factors_to_reuse.append(factor)
self.time_step_factors_to_reuse.append (self.current_num_time_steps)
\# adapt the num_observed attribute of the observed part
self.symbols_info[symbol] ["num_observed"] += 1
\# adapt the weights for updating rotation and translation
self.update_weights("translation")
self.update_weights("rotation")
def optimize_sensor_pose(self, initial, initial_guess_new_sensor):
"""
Find the pose of the sensor in the world frame that fits the observations and the assumed model best. The initial guess is the estimated pose of the sensor in the world frame at the beginning of the optimization.

Parameters
initial : gtsam.Values
Initial values of the variables except the new sensor.
initial_guess_new_sensor : gtsam.Pose3
Initial guess for the pose of the sensor in the world frame.

Returns
best_pose_new_sensor : gtsam.Pose3
Pose of the sensor in the world frame that fits the observations best.
best_error : float
Optimization error of the best pose of the sensor in the world frame.
result: gtsam.Values
Optimized values of all variables.
"""
\# get the current sensor symbol
symbol_new_sensor = self.get_sensor_symbol(self.current_num_time_steps)
\# insert the initial guess into the current result
initial.insert (symbol_new_sensor, initial_guess_new_sensor)
\# optimize the graph with the current result
result \(=\) self.optimize_LM(initial)
error \(=\) self.graph.error (result)
new_pose_new_sensor = result.atPose3(symbol_new_sensor)
return new_pose_new_sensor, error, result
def add_pose_to_trajectory(self, pose):
"""
Add the estimated pose of the sensor in the world frame to the sensor trajectory and to the symbols_info dictionary, increase the number of time steps by one, and update the default construction noise.

Parameters
pose : gtsam.Pose3
Estimated pose of the sensor in the world frame.
"""
\# add the pose to the sensor trajectory
self.sensor_trajectory.append (pose)
\# add a new entry to the symbols_info dictionary for the sensor in the new time step
self.symbols_info[self.get_sensor_symbol(self.current_num_time_steps)] = \{
"num_observed": 0,
"pose_in_world_frame": \{
"x": pose.translation() [0],
"y": pose.translation() [1],
"z": pose.translation() [2], "rotation_matrix": pose.rotation(),
\},
"dependency_groups": \{\},
"groups_with_degrees_of_freedom": \{\},
"std_degrees_of_freedom": \{\},
\}
\# increase the number of time steps by one
self.current_num_time_steps += 1
\# adapt construction noise
self.construction_noise_rotation_default *= self.updating_factor_noise
self.construction_noise_translation_default \(*=\) self.updating_factor_noise
```

Additional functions, not shown here:
def get_initial_values(self)
"""
\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#

# Factor graph

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```
def build_construction_noise_dictionary(self):
    """
    Build the construction noise dictionary. This function is called once in the beginning such
    that the dictionary does not have to be rebuilt in each time step. The keys of the dictionary
    are tuples of two symbols, e.g. (self.w1, self.L2). The values are dictionaries containing the
    construction noise for each axis,
    e.g. \(\{" x ": 0.1, \quad " y ": 0.1, ~ " z ": 0.1, ~ " r o l l ": ~ 0.1, ~ " p i t c h ": ~ 0.1, ~ " y a w ": ~ 0.1\} . ~\)
    " 1 "
    \# define which variables are connected by factors
    self.define_used_factors()
    \# the order of the symbols stays the same, so we do not have to check each pair twice
    for tuple_of_symbols in self.construction_noise_dictionary.keys():
        first_symbol = tuple_of_symbols[0]
        second_symbol = tuple_of_symbols[1]
        \# initialize the construction noise with "default"
        \# this stays the same if the parts are not dependent on each other
        construction_noise \(=\) \{
            'roll": "default",
            "pitch": "default",
            "yaw": "default",
            "x": "default",
            "y": "default",
            "z": "default",
        \}
        \# adapt the construction noise according to the dependency groups
        for axis in self.symbols_info[first_symbol]["dependency_groups"].keys():
            if second_symbol in self.symbols_info[first_symbol]["dependency_groups"] [axis]:
                construction_noise[axis] = "dependent"
        \# adapt the construction noise according to the degrees of freedom
        for axis in self.symbols_info[first_symbol]["groups_with_degrees_of_freedom"]:
            new_construction_noise = self.get_construction_noise_with_degrees_of_freedom(
                first_symbol, second_symbol, axis
            )
            construction_noise[axis] = new_construction_noise
        \# add the construction noise to the dictionary
        self.construction_noise_dictionary[tuple_of_symbols] = copy.deepcopy(construction_noise)
def define_used_factors(self):
    """
    Define which parts of the truck should be connected with factors.
    The variables have a fixed order, so each combination of symbols appears exactly once.
    Define the construction noise dictionary.
    """
    \# define the keys in the form of a dictionary (if w1 is in factors [g0], [(g0, w1)] is a key)
    factors \(=\{\)
        self.g0: [self.l1, self.w1, self.w2, self.G0],
        self.G0: [self.L1, self.W1, self.W2],
        self.w1: [self.l1, self.W2, self.W1],
        self.w2: [self.w3, self.W2],
        self.w3: [self.l2, self.W3],
        self.W1: [self.L1, self.W2],
        self.W2: [self.W3],
        self.W3: [self.L2],
        self.l1: [self.L1],
        self.l2: [self.L2],
        self.L1: [],
        self.L2: [],
    }
    # build the structure of the construction noise dictionary
    for symbol in factors.keys():
        for other_symbol in factors[symbol]:
        self.construction_noise_dictionary[(symbol, other_symbol)] = {}
def create_factor_graph(self):
    """
    (Re-)Create the factor graph and add the prior and construction factors to the graph.
    """
    # create an empty graph
    self.graph = gtsam.NonlinearFactorGraph()
    # the world origin is exactly at the start of the left loading edge
    # therefore, we set the respective noise to zero
    world_origin_noise = gtsam.noiseModel.Diagonal.Sigmas(np.array([0] * 6))
    # define the prior factor and add it to the graph
    factor_prior = gtsam.PriorFactorPose3(self.g0, gtsam.Pose3(), world_origin_noise)
    self.graph.add(factor_prior)
    # add the construction factors
    self.add_construction_factors()
def add_construction_factors(self):
    Function to add the construction factors to the graph. Since these factors depend on
    the model of the truck, we have to add them after each time step (after recreating the graph).
    The values of the BetweenFactorPose3 are based on the transformation between the poses of
    two parts in the world frame (as stated in symbols_info). The noise model is based on
    the construction noise dictionary.
    """
    # define the factors between all pairs of parts (order as stated in self.symbols_parts)
    for tuple_of_symbols in self.construction_noise_dictionary.keys():
        first_symbol = tuple_of_symbols[0]
        second_symbol = tuple_of_symbols[1]
        # get the transformation between the two parts
        trafo_first_to_second_symbol = self.symbols_info_to_pose_in_world_frame(
            first_symbol
        ).between(self.symbols_info_to_pose_in_world_frame(second_symbol))
        construction_noise_model = copy.deepcopy(
            self.construction_noise_dictionary[tuple_of_symbols]
        )
        # substitute the entries in the construction noise model by their corresponding values
        for axis in construction_noise_model.keys():
            if construction_noise_model[axis] == "default":
                if axis in ["x", "y", "z"]:
                    construction_noise_model[axis] = self.construction_noise_translation_default
                    else:
                    construction_noise_model[axis] = self.construction_noise_rotation_default
                elif construction_noise_model[axis] == "dependent":
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                if axis in ["x", "y", "z"]:
                    construction_noise_model[axis] = self.construction_noise_translation_dependent
        else:
                            construction_noise_model[axis] = self.construction_noise_rotation_dependent
        # if it is not "default" or "dependent", it gives the additional noise for this factor
        else:
            if axis in ["x", "y", "z"]:
            construction_noise_model[axis] += self.construction_noise_translation_default
        else:
            construction_noise_model[axis] += self.construction_noise_rotation_default
        # get the noise model for the factor
        construction_noise_model = self.get_noise_model(**construction_noise_model)
        # define the factor and add it to the graph
        factor = gtsam.BetweenFactorPose3(
            first_symbol, second_symbol, trafo_first_to_second_symbol, construction_noise_model
        )
        self.graph.add(factor)
def update_construction_noise(self):
    Update the default construction noise for rotation and translation according to the number of
    steps. The construction noise follows the function
        c(n) = lambda^n * (c(0) - lower_bound) + lower_bound,
    thus we compute the new construction noise as
        c(n+1) = lambda * c(n) + (1-lambda)*lower_bound.
    """
    self.construction_noise_translation_default = (
        self.updating_factor_noise * self.construction_noise_translation_default
        + (1 - self.updating_factor_noise) * self.lower_bound_construction_noise
    )
    self.construction_noise_rotation_default = (
        self.updating_factor_noise * self.construction_noise_rotation_default
        + (1 - self.updating_factor_noise) * self.lower_bound_construction_noise
    )
def update_weights(self, update_type):
    """
    Update the weights of a part according to how often it has been observed.
    The weights follow the function
        w(n) = lambda^n * (w(0) - lower_bound) + lower_bound,
    thus we compute the new weight as
        w(n+1)=lambda*w(n) + (1-lambda)*lower_bound.
    Parameters
    update_type : str
        Either "translation" or "rotation".
    """
    if update_type == "translation":
        self.weight_translation = (
            self.updating_factor_weights * self.weight_translation
            + (1 - self.updating_factor_weights) * self.lower_bound_weights
        )
    elif update_type == "rotation":
        self.weight_rotation = (
            self.updating_factor_weights * self.weight_rotation
            + (1 - self.updating_factor_weights) * self.lower_bound_weights
        )
```

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```
def optimize_LM(self, initial):
```

def optimize_LM(self, initial):
"""
Optimize the factor graph using Levenberg-Marquardt optimization of gtsam.
Parameters
initial : gtsam.Values
Initial values of all variables.
Returns
result : gtsam.Values
Optimized values of all variables, including the estimated sensor pose.
"""
\# define the optimizer
params = gtsam.LevenbergMarquardtParams()
params.setVerbosityLM("ERROR")
optimizer = gtsam.LevenbergMarquardtOptimizer(self.graph, initial, params)
\# optimize the graph
result = optimizer.optimize()
return result
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# Helper functions

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def get_construction_noise_with_degrees_of_freedom(
self, first_symbol, second_symbol, axis_first_symbol
):
"""
Get the the construction noise for a part in a group with degree of freedom to other parts.
Parameters
first_symbol : gtsam.Symbol
Currently investigated symbol.
second_symbol : gtsam.Symbol
Symbol of the part we want to investigate the connection to first_symbol with.
axis_first_symbol : str
Respective axis for the group with an degree of freedom.
Returns
construction_noise : str or float
"dependent" if the two symbols are in the same group,
otherwise the additional standard deviation for the noise is returned.
"""
if (
second_symbol
in self.symbols_info[first_symbol]["groups_with_degrees_of_freedom"][axis_first_symbol]
):
return "dependent"
else:
return self.symbols_info[first_symbol]["std_degrees_of_freedom"][axis_first_symbol]
def get_noise_model(self, roll=0.1, pitch=0.1, yaw=0.1, x=0.3, y=0.3, z=0.3, rot=None, pos=None):
"""
Get a noise model for the observations.
The noise model is a diagonal matrix with the given standard deviations as entries.
Parameters

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roll : float, optional
Standard deviation of the noise in the rotation around the \(x\)-axis. The default is 0.1.
pitch : float, optional
Standard deviation of the noise in the rotation around the \(y\)-axis. The default is 0.1.
yaw : float, optional
Standard deviation of the noise in the rotation around the \(z\)-axis. The default is 0.1.
\(x\) : float, optional
Standard deviation of the noise in the \(x\)-coordinate. The default is 0.3.
\(y\) : float, optional
Standard deviation of the noise in the \(y\)-coordinate. The default is 0.3.
\(z\) : float, optional
Standard deviation of the noise in the z-coordinate. The default is 0.3.
rot : float, optional
Standard deviation of the noise in the rotation around all axes.
If rot is given, roll, pitch and yaw are set to rot. The default is None.
pos : float, optional
Standard deviation of the noise in the translation in all directions.
If pos is given, \(x, y\) and \(z\) are set to pos. The default is None.

Returns
noise_model : gtsam.noiseModel
Noise model for the observations.
"""
if rot is not None:
roll, pitch, yaw \(=\) rot, rot, rot
if pos is not None:
\(x, y, z=p o s\), pos, pos
return gtsam.noiseModel.Diagonal. Sigmas(np.array([roll, pitch, yaw, x, y, z]))
```

def get_rotation_matrix(self, angle, axis="x"):

```

Get the rotation matrix for a rotation around the \(x, y\), or \(z\) axis.

Parameters
angle : float
Angle of the rotation in radians.
axis : str, optional
Axis of the rotation (" \(x\) ", " \(y\) ", or " \(z\) "). The default is " \(x\) ".

Returns
rotation_matrix : np.array
Rotation matrix.
"""
if axis == "x" or axis == "roll":
return np.array (
[
\([1,0,0]\),
[0, np.cos(angle), -np.sin(angle)],
[0, np. \(\sin (\) angle), np.cos(angle)],
]
)
elif axis == "y" or axis == "pitch":
return np.array (
[
[np.cos(angle), 0, np.sin(angle)],
[0, 1, 0],
[-np.sin(angle), 0, np.cos(angle)],
]
```

        elif axis == "z" or axis == "yaw":
        return np.array(
            [
            [np.cos(angle), -np.sin(angle), 0],
            [np.sin(angle), np.cos(angle), 0],
            [0, 0, 1],
            ]
        )
    else: # incorrect input
        print("Incorrect input for axis, please use x, y, or z!")
        return None
    def get_joint_rotation_matrix(self, angles, axes="xyz"):
Get the rotation matrix for a rotation around the }x,y\mathrm{ , and/or z axis, angle in radians.
Parameters
angles : list of float
Angles of the rotation in radians around the axes specified in "axes".
axes : str, optional
Axes of the rotation (sequence of "x", "y", and "z"). The default is "xyz".
Returns
rotation_matrix : np.array
Rotation matrix.
"""
rotation_matrix = np.eye(3)
for i in range(len(angles)):
rotation_matrix = np.dot(self.get_rotation_matrix(angles[i], axes[i]), rotation_matrix)
return rotation_matrix
"""
Additional functions, not shown here:
def get_dominant_part(self, list_of_parts)
def get_width_loading_platform(self)
def get_current_sensor_pose(self)
-
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# Visualization

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"""
Visualization functions, not shown here:
def plot_sensor_trajectory(...)
def plot_truck(...)
def create_cuboid(...)
def create_cylinder(...)
def create_coordinate_frame(...)
"""

```

\section*{\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#\#}
```


# Exemplary Test

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```
```

def test_model_error_construction_noise(

```
def test_model_error_construction_noise(
    data_file, obs_noise_sd, constr_noise_sd, parameter_values, repeat_test
    data_file, obs_noise_sd, constr_noise_sd, parameter_values, repeat_test
):
```

):

```
1276 Test how fast and well the model of the truck is updated in dependence of the construction noise.
Parameters
data_file : pickle file
    Pickle file containing the observed data in form of a dictionary.
obs_noise_sd : float
    Standard deviation of the additional artificial noise for the observations.
constr_noise_sd : float
    Standard deviation of the additional artificial noise for the construction of the truck.
parameter_values : list of float
    Different values of the construction noise used for testing.
repeat_test : int
    States how often the test should be repeated.
Returns
means : ndarray of shape (len(parameter_values), num_timesteps)
    Means of the model error for the different parameter values of all test trials.
std_devs : ndarray of shape (len(parameter_values), num_timesteps)
    Standard deviations of the model error for the different parameter values of all test trials.
"""
\# set a seed to get reproducible results
np.random.seed(11)
\# get the observations from the data file
observation_list = data_to_observations(data_file)
\# basic parameter settings
observation_noise_translation_default, observation_noise_rotation_default = 0.17, 0.14
construction_noise_rotation_default = 1.5
updating_factor_noise, updating_factor_weights = 0.94, 0.97
weight_translation, weight_rotation \(=0.80,0.80\)
align_every_n_steps = 32
\# create a basic instance of the class Truck3D with the parameters defined above
Truck_basic = Truck3D(
    observation_noise_translation_default=observation_noise_translation_default,
    observation_noise_rotation_default=observation_noise_rotation_default,
    construction_noise_rotation_default=construction_noise_rotation_default,
    updating_factor_noise=updating_factor_noise,
    updating_factor_weights=updating_factor_weights,
    initial_weights_translation=weight_translation,
    initial_weights_rotation=weight_rotation,
    align_every_n_steps=align_every_n_steps,
)
    \# list that will contain the model errors for all test trials
    results = []
    for _ in range(repeat_test):
    \# copy the instance of the Truck3D class
    Truck_imprecise_configuration = copy.deepcopy(Truck_basic)
    \# define an imprecise truck configuration with additional random noise
    constr_noise = np.random.normal(0, constr_noise_sd, 11)
    Truck_imprecise_configuration = get_imprecise_truck_configuration(
            Truck_imprecise_configuration, constr_noise
    )
    \# list that will contain the model errors for one test trial
    model_errors = []
```

    # add random noise to the observations
    noisy_observations = add_noise_to_data(observation_list, obs_noise_sd)
    # run the sensor pose estimations for all parameter values
    for j in range(len(parameter_values)):
        # copy the imprecise Truck3D object
        Truck = copy.deepcopy(Truck_imprecise_configuration)
        # set the value of the construction noise
        parameter = parameter_values[j]
        Truck.construction_noise_translation_default = parameter
        # run the sensor pose estimation on these observations
        model_error, _ = simulate_sensor_trajectory_model_error(Truck, noisy_observations)
        model_errors.append(model_error)
        results.append(model_errors)
    means = np.mean(results, axis=0)
    std_devs = np.std(results, axis=0)
    return means, std_devs
    def data_to_observations(data_file):
Get the observations saved in a data file.
Parameters
data_file : pickle file
Opened pickle file containing the data.
Returns
all_observations : list of dict
List containing the observations for each time step.
""""
data_dict = pickle.load(data_file)
all_observations = []
for frame in data_dict.values():
current_observation = {}
for symbol in frame.keys():
pos, rot = frame[symbol]["pos"], frame[symbol]["rot"]
current_observation[symbol] = gtsam.Pose3(gtsam.Rot3(rot), pos)
all_observations.append(current_observation)
return all_observations
def add_noise_to_data(data, noise_sd):
"""
Add random additional noise to data.
Parameters
data : list of dict
Data of gtsam poses, where we want to add noise.
noise_sd : float

```

\section*{B. Code Pose Estimation}
```

Standard deviation of the normally distributed noise.
Returns
noisy_data : list of dict
Data of gtsam poses with random noise.
"""
noisy_data $=$ copy. deepcopy (data)
for frame in noisy_data:
for symbol_string in frame.keys():
\# get random numbers for the noise of position and orientation
noise_pos = np.random.normal(0, noise_sd, 3)
noise_rot $=$ np.random.normal(0, min(noise_sd, 0.05), 3)
\# create a rotation matrix from the local coordinates noise_rot of $S O$ (3)
noise_rot_matrix = local_update_SO3(np.eye(3), noise_rot)
\# transform the noise to a gtsam pose
noise_pose $=$ gtsam. Pose3(gtsam.Rot3(noise_rot_matrix), noise_pos)
\# add the noise to the data
frame[symbol_string] = frame[symbol_string].compose(noise_pose)
return noisy_data
def local_update_SO3(rotation_matrix, vector):
"""
Local update on $S O(3)$ using the hat-operator and the matrix exponential as a retraction.
Parameters
rotation_matrix : ndarray of shape $(3,3)$
Original rotation matrix $R O$.
vector : ndarray of shape (3,)
Vector $x i$ of the local coordinates on $S O(3)$.
Returns
ndarray of shape $(3,3)$
Local update of the original rotation matrix $R O$.
"""
return rotation_matrix @ matrix_exp(vector)
def matrix_exp(vector):
"""
Matrix exponential for a skew-symmetric matrix.
Uses the hat-operator on $S O$ (3) and computes the resulting matrix with Rodrigues' forumla.
Parameters
----------
vector: ndarray of shape (3,)
Vector $x i$ of the local coordinates on $S O(3)$.
Returns
$\exp$ (hat(vector)) : ndarray of shape $(3,3)$
The matrix exponential of hat(vector).
"""
theta $=\mathrm{np} . \operatorname{linalg}$. norm(vector)
if theta $==0$ :
return np.eye(3)

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```

    else:
        return (
            np.eye(3)
            + np.sin(theta) / theta * hat_operator(vector)
            + (1 - np.cos(theta)) / theta**2 * hat_operator(vector) @ hat_operator(vector)
        )
    def hat_operator(vector):
N"
Implementation of the hat-operator for SO(3).
Parameters
vector : ndarray of shape (3,)
Vector xi of the local coordinates on SO(3).
Returns
hat(vector) : ndarray of shape (3,3)
Skew-symmetric matrix, output of the hat-operator.
"""
return np.array(
[
[0, -vector[2], vector[1]],
[vector[2], 0, -vector[0]],
[-vector[1], vector[0], 0],
]
)
def simulate_sensor_trajectory_model_error(Truck, observations):
"""
Simulate a run of the sensor pose estimation for given observations.
Parameters
Truck : Truck3D
Instance of the Truck3D class with the respective parameters.
observations : list of dict
List of the noisy observations.
Returns
model_error : list of float
List of the model errors for each time step.
"""
\# list containing the model errors for each time step
model_error = []
\# get the initial model error
model_error.append(evaluate_model_error(Truck))
for i in range(len(observations)):
\# estimate the sensor pose
_, _, result = Truck.estimate_sensor_pose(observations[i])
\# update the truck configuration according to the results of the previous estimation
Truck.update_truck_configuration(result)
\# evaluate the current error of the estimated model
model_error.append(evaluate_model_error(Truck))

```

return model_error
    Returns
    cumulated_distance : float
        Model error cumulated for all parts of the truck.
    "॥"
    cumulated_distance \(=0\)
    for symbol in Truck.symbols_info:
        if symbol not in Truck.symbols_sensor:
            \# compute the distance between the model and the ground truth
            pose_difference = Truck.symbols_info_to_pose_in_world_frame (
                symbol, true_or_estimated="estimated"
            cumulated_distance += pose_norm(pose_difference)
    return cumulated_distance
def pose_norm(pose):
    "" "
    Parameters
    pose : gtsam.Pose3
        Pose to compute the norm.
    Returns
    norm : float
        Norm of the given pose.
    """
    pose_norm_position \(=\) np.linalg.norm(pose.translation())
    \(\mathrm{R}=\) pose.rotation().matrix()
    I = np.identity(3)
    pose_norm_rotation \(=\) np.linalg.norm \((I-R)\)
    return pose_norm_position + pose_norm_rotation
    visualization functions
" " " "
```

def evaluate_model_error(Truck):
Evaluate the model error of the truck. The model error is defined as the distance between
the poses of the parts in the model and the poses of the parts in the ground truth.
The distance in the position is computed as the norm of the difference between the positions.
The distance in the rotation R1, R2 is computed as the Frobenius norm of I-R1R2^T.
Parameters
Truck : Truck3D
Truck, where we look for the error between the assumed model and the ground truth.

```
            ). between(Truck. symbols_info_to_pose_in_world_frame(symbol, true_or_estimated="true"))
    Norm of the pose used to measure the model error. The norm of the position is computed as the
    Euclidean norm. The norm of the rotation \(R\) is computed as the Frobenius norm of \(I-R\).

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