

DIPLOMARBEIT

# Erkennung und 3D Posenschätzung Zusammengesetzter Objekte

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durch

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DIPLOMA THESIS

# Composite Object Detection and 3D Pose Estimation

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supervised by

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## Kurzfassung

Die orthogonale Gruppe  $O_n$  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  $SO_n$  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  $SE_n$  besteht aus allen Paaren (R, t), wobei R eine Rotation in  $SO_n$  und t ein Vektor in  $\mathbb{R}^n$  ist. Die Elemente von  $SE_n$  können die Posen von Objekten repräsentieren. Diese drei Untergruppen von  $GL_n$  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 TM einer glatten Mannigfaltigkeit M auf M, die bestimmte Eigenschaften hat. Auf SO<sub>n</sub> und SE<sub>n</sub> 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  $O_n$  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  $SO_n$  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  $SE_n$  comprises all pairs (R, t), where R is a rotation in  $SO_n$  and t is a translation vector in  $\mathbb{R}^n$ . An element of  $SE_n$  can be used to represent the pose of an object. These three subgroups of  $GL_n$  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 TM 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 SO<sub>n</sub> and SE<sub>n</sub>. 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

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Florendie



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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 SE<sub>3</sub>. It is commonly referred to as the position and orientation of this object. A matrix  $T \in SE_3$  has the form

$$T = \begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \in \mathbb{R}^{4 \times 4}$$

for a vector  $t \in \mathbb{R}^3$  indicating the position and a rotation matrix  $R \in SO_3$  indicating the orientation. The special orthogonal group  $SO_3$  is the subgroup of all matrices in  $O_3$  with determinant 1. The orthogonal group  $O_3$  is the group of all matrices  $A \in \mathbb{R}^{3\times 3}$  with

$$AA^T = A^T A = I_3$$

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  $h: SE_3 \to \mathbb{R}^n$  that predicts measurements for given poses, we search for the matrix  $T \in SE_3$  that best approximates the measurements z under the function h. Hence, we have to solve

$$\underset{T \in SE_3}{\arg\min} \|h(T) - z\|.$$

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

$$\begin{pmatrix} R_1 & t_1 \\ 0 & 1 \end{pmatrix} + \begin{pmatrix} R_2 & t_2 \\ 0 & 1 \end{pmatrix} = \begin{pmatrix} R_1 + R_2 & t_1 + t_2 \\ 0 & 2 \end{pmatrix}$$

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of two matrices  $T_1$  and  $T_2$  in SE<sub>3</sub> is not in SE<sub>3</sub> anymore. Even the sum  $R_1 + R_2$  of two rotation matrices  $R_1, R_2 \in SO_3$  is in general not in SO<sub>3</sub>. Therefore, we cannot expect to receive a new valid estimate  $T^{(t+1)} \in SE_3$  by adding some matrix  $\delta^{(t)}$  to  $T^{(t)} \in SE_3$ . To work around this problem, we will exploit the structure of SO<sub>3</sub> and SE<sub>3</sub> as smooth manifolds and use retractions. With the exponential map

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

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

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 SE<sub>3</sub>.

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  $SO_3$  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  $SO_3$ . 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  $O_n$ ,  $SO_n$ , and  $SE_n$ . 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  $SO_n$ , 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  $SO_n$ .

#### 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  $O_n$  and the special orthogonal group  $SO_n$  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  $O_n$  and  $SO_n$  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*  $GL_n(\mathbb{R})$ , or from now on just  $GL_n$ , 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  $GL_n$ , we can define other matrix groups.

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

$$AA^T = A^T A = I.$$

We have to check, whether the name group is justified for  $O_n$ . In the following theorem, some basic properties of  $O_n$  are investigated.

**Theorem 2.1.2** (Properties of  $O_n$ ). Let n be a positive natural number. Then the following properties of the orthogonal group  $O_n$  hold.

- (i) The orthogonal group  $O_n$  is a subgroup of  $GL_n$ .
- (ii) The column vectors of any matrix  $A \in O_n$  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  $A \in O_n$  have (Euclidean) norm 1.
- (iv) The column vectors of any matrix  $A \in O_n$  form an orthonormal basis of  $\mathbb{R}^n$ .

- (v) For all  $A \in O_n$  the determinant det A is either +1 or -1.
- (vi) Any  $A \in O_n$  preserves the inner product on  $\mathbb{R}^n$ , i.e.  $\langle x, y \rangle = \langle Ax, Ay \rangle$ .
- (vii) If  $\lambda$  is an eigenvalue of an  $A \in O_n$ , then  $|\lambda| = 1$  holds.
- *Proof.* (i) First, since for all  $A \in O_n$  the property  $AA^T = A^T A = I$  holds, the transposed matrix  $A^T$  is the inverse of A. Therefore, A is regular and  $O_n \subseteq GL_n$ .

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

$$(AB)(AB)^{T} = ABB^{T}A^{T} \stackrel{B \in \mathcal{O}_{n}}{=} AIA^{T} \stackrel{A \in \mathcal{O}_{n}}{=} I$$

shows the closure of  $O_n$ . Also, matrix multiplication is associative.

Thus,  $O_n$  is a group and because of  $O_n \subseteq GL_n$  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  $A \in O_n$  translates to

$$\begin{pmatrix} 1 & 0 & \cdots & 0 \\ 0 & 1 & \cdots & 0 \\ \vdots & \vdots & \ddots & \vdots \\ 0 & 0 & \cdots & 1 \end{pmatrix} = I = A^T A = \begin{pmatrix} \langle A_1, A_1 \rangle & \langle A_1, A_2 \rangle & \cdots & \langle A_1, A_n \rangle \\ \langle A_2, A_1 \rangle & \langle A_2, A_2 \rangle & \cdots & \langle A_2, A_n \rangle \\ \vdots & \vdots & \ddots & \vdots \\ \langle A_n, A_1 \rangle & \langle A_n, A_2 \rangle & \cdots & \langle A_n, A_n \rangle \end{pmatrix}.$$
 (2.1)

Thus, the product  $\langle A_i, A_j \rangle$  for  $i \neq j$  equals 0, so the column vectors of A are pairwise orthogonal.

- (iii) From (2.1) directly follows  $\langle A_i, A_i \rangle$  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  $A \in O_n$  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  $A \in O_n$  form an orthonormal basis of  $\mathbb{R}^n$ .
- (v) Since the determinant of a matrix is compatible with matrix multiplication and transposition, the equation  $I = AA^T$  leads to

$$1 = \det(I) = \det(AA^T) = \det(A)\det(A^T) = \det(A)^2$$

and thus  $det(A) = \pm 1$  for  $A \in O_n$ .

(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  $\langle Ax, Ay \rangle = (Ax)^T (Ay) = x^T A^T Ay = x^T y = \langle x, y \rangle$ .

(vii) The eigenvalues of  $A \in O_n$  are the solutions  $\lambda$  of the equation  $Av = \lambda v$ . 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  $\mathbb{C} \setminus \mathbb{R}$ ) and  $A^T = A^{-1}$  and get

$$\|Av\|^{2} = \langle Av, Av \rangle$$
  
=  $\overline{(Av)}^{T} (Av)$   
=  $\overline{v}^{T} A^{T} Av$   
=  $\overline{v}^{T} v$   
=  $\langle v, v \rangle = \|v\|^{2}$ .

Thus,  $||v|| = ||Av|| = ||\lambda v|| = |\lambda| ||v||$  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  $O_n$  on any vector space V with an inner product  $\langle \cdot, \cdot \rangle$  on V: The orthogonal group  $O_n$  is the set of all automorphisms  $f: V \to V$  that preserve the inner product, i.e.  $\langle v, w \rangle = \langle f(v), f(w) \rangle$  [22, §0].

The automorphism  $\mathbb{R}^n \to \mathbb{R}^n$  represented by a matrix  $A \in O_n$  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

$$\begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix} \in \mathcal{O}_2$$

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

$$\begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} \in \mathcal{O}_2$$

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

$$\begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix} = \begin{pmatrix} -1 & 0 \\ 0 & 1 \end{pmatrix} \in \mathcal{O}_2$$

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  $SO_n$  is the set of all matrices  $A \in O_n$  with det(A) = 1.

**Theorem 2.1.6.** The special orthogonal group  $SO_n$  is a subgroup of  $O_n$ .

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

$$\det(AB) = \det(A)\det(B) = 1.$$

Thus,  $SO_n$  is a group and because of  $SO_n \subseteq O_n$  a subgroup of the orthogonal group.  $\Box$ 

**Lemma 2.1.7.** For an odd n > 1, all  $A \in SO_n$  have 1 as an eigenvalue.

*Proof.* The eigenvalues of a matrix  $A \in SO_n$  are the zeros of the characteristic polynomial  $\chi_A(\lambda) = \det(A - \lambda I)$ . It holds with  $A^T = A^{-1}$ ,  $\det(A) = 1$ , and  $\det(B) = \det(B^T)$  for all matrices  $B \in \mathbb{R}^{n \times n}$ 

$$det(A - I) = det(A - AA^{-1})$$
  
= det(A(I - A^{-1}))  
= det(A) det(I - A^{T})  
= (-1)^n det(A^T - I)  
= (-1)^n det((A^T - I)^T)  
= (-1)^n det(A - I).

For an odd n, the equation reads as det(A - I) = -det(A - I), so 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

$$\begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} \in \mathrm{SO}_2$$

shows.

The special orthogonal group  $SO_n$  is also referred to as the rotation group. Geometrically, the matrices in  $SO_2$  represent planar rotations around the origin, and the matrices in  $SO_3$  represent rotations in  $\mathbb{R}^3$  around an axis through the origin [26, Chapter 1].

*Example* 2.1.8 (SO<sub>2</sub>). To get a matrix  $A \in SO_2$ , the two column vectors  $(a_{11}, a_{21})^T$  and  $(a_{12}, a_{22})^T$  of A have to be orthogonal, so

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

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

$$\begin{pmatrix} a_{12} \\ a_{22} \end{pmatrix} = \pm \begin{pmatrix} -a_{21} \\ a_{11} \end{pmatrix}.$$

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  $\{\arccos a_{11}, -\arccos a_{11}\}$  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

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

holds.

We conclude that all matrices  $A \in SO_2$  are of the form

$$A = \begin{pmatrix} \cos \alpha & -\sin \alpha \\ \sin \alpha & \cos \alpha \end{pmatrix}$$

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 SO<sub>2</sub> and the respective  $\alpha$ .

As seen in Example 2.1.8, any element of SO<sub>2</sub> 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 SO<sub>2</sub> has one *degree of freedom*.

**Lemma 2.1.9.** Applying two rotations from  $SO_2$  by angles  $\alpha$  and  $\beta$  is the same as applying one rotation from  $SO_2$  by the angle  $\alpha + \beta$ .

Especially, the group  $SO_2$  with the usual matrix multiplication is commutative.

*Proof.* Let  $A, B \in SO_2$  be two matrices in the special orthogonal group. We have seen that there exist  $\alpha, \beta \in (-\pi, \pi]$  such that

$$A = \begin{pmatrix} \cos \alpha & -\sin \alpha \\ \sin \alpha & \cos \alpha \end{pmatrix} \text{ and } B = \begin{pmatrix} \cos \beta & -\sin \beta \\ \sin \beta & \cos \beta \end{pmatrix}.$$

Multiplying these two matrices yields

$$AB = \begin{pmatrix} \cos \alpha & -\sin \alpha \\ \sin \alpha & \cos \alpha \end{pmatrix} \begin{pmatrix} \cos \beta & -\sin \beta \\ \sin \beta & \cos \beta \end{pmatrix}$$
$$= \begin{pmatrix} \cos \alpha \cos \beta - \sin \alpha \sin \beta & -\cos \alpha \sin \beta - \sin \alpha \cos \beta \\ \sin \alpha \cos \beta + \cos \alpha \sin \beta & -\sin \alpha \sin \beta + \cos \alpha \cos \beta \end{pmatrix}.$$

Using sum identities for trigonometric functions, results in

$$AB = \begin{pmatrix} \cos(\alpha + \beta) & -\sin(\alpha + \beta) \\ \sin(\alpha + \beta) & \cos(\alpha + \beta) \end{pmatrix}$$

which is the matrix for a rotation by the angle  $\alpha + \beta$ .

Analogous computations for the matrix product BA give the same result which shows commutativity.

The elements of SO<sub>3</sub>, the group of rotations in  $\mathbb{R}^3$  [26, Chapter 1], do not behave as nicely as those of SO<sub>2</sub>. 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 SO<sub>2</sub> we have derived above [32, p. 12.4.14]:

rotation around the $x$ -axis	rotation around the $y$ -axis	rotation around the $z$ -axis

/1	0	0	$\cos \alpha$	0	$-\sin\alpha$		$\cos \alpha$	$-\sin \alpha$	0)
0	$\cos \alpha$	$-\sin \alpha$	0	1	0		$\sin \alpha$	$\cos \alpha$	0
$\setminus 0$	$\sin \alpha$	$\cos \alpha$	$\sin \alpha$	0	$\cos \alpha$	/	0	0	1/

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  $SO_3$  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  $E_n$  is the set of all matrices of the form

$$\begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix}$$

where R is a matrix in  $O_n$  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  $\mathbb{E}_n$  are in  $\mathbb{R}^{(n+1)\times(n+1)}$ . A translation  $\mathbb{R}^n \to \mathbb{R}^n$  is not a linear transformation, thus it cannot be represented by an  $(n \times n)$ -matrix. Working with homogeneous coordinates  $(x^T, 1)^T$  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  $SE_n$  is the set of all matrices of the form

$$\begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \tag{2.2}$$

where R is a rotation matrix in  $SO_n$  and t is a translation vector in  $\mathbb{R}^n$ .

This set is also called the set of proper rigid transformations [46].

Since  $SO_n \subseteq O_n$ , it holds that  $SE_n \subseteq E_n$ . Furthermore, the following theorem holds.

**Theorem 2.1.12.** The Euclidean group  $E_n$  and the special Euclidean group  $SE_n$  are subgroups of the general linear group  $GL_{n+1}$ .

*Proof.* The proof for  $E_n$  and  $SE_n$  are completely analogous, so only the proof for  $SE_n$  is given.

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

$$\det \begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} = \det(R) \det(1) = 1.$$

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

We need to verify the group axioms for  $SE_n$  next. For  $R = I_n \in SO_n$  and t being the zero vector in  $\mathbb{R}^n$ , the matrix of the form (2.2) is the identity element in  $GL_{n+1}$ . Let

$$A_1 = \begin{pmatrix} R_1 & t_1 \\ 0 & 1 \end{pmatrix} \quad \text{and} \quad A_2 = \begin{pmatrix} R_2 & t_2 \\ 0 & 1 \end{pmatrix}$$

be elements of  $SE_n$ . Then

$$A_1 A_2 = \begin{pmatrix} R_1 R_2 & R_1 t_2 + t_1 \\ 0 & 1 \end{pmatrix}$$

which is in SE<sub>n</sub>, since the product  $R_1R_2$  of two rotation matrices is again a rotation matrix and  $R_1t_2 + t_1 \in \mathbb{R}^n$ . If we want  $A_2$  to be the inverse matrix of  $A_1$ , then  $R_1R_2 = I$  and  $R_1t_2 + t_1 = 0$  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

$$A_2A_1 = \begin{pmatrix} R_1^{-1} & -R_1^{-1}t_1 \\ 0 & 1 \end{pmatrix} \begin{pmatrix} R_1 & t_1 \\ 0 & 1 \end{pmatrix} = \begin{pmatrix} R_1^{-1}R_1 & R_1^{-1}t_1 - R_1^{-1}t_1 \\ 0 & 1 \end{pmatrix} = I_{n+1}.$$

As a result,  $SE_n$  is a group and altogether a subgroup of  $GL_{n+1}$ .

We can uniquely identify the elements of  $SE_n$  with pairs in  $SO_n \times \mathbb{R}^n$  if needed:

$$\begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \quad \longleftrightarrow \quad (R, t).$$

$$(2.3)$$

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

$$T: \begin{cases} \mathbb{R}^n \to \mathbb{R}^n \\ x \mapsto Rx + t \end{cases}$$
(2.4)

For an element of SE<sub>3</sub>, one can choose a rotation matrix  $R \in SO_3$  which has three degrees of freedom, and a translation vector  $t \in \mathbb{R}^3$  where three values can be chosen independently. In total, SE<sub>3</sub> has six degrees of freedom.

To illustrate SO<sub>3</sub>, 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  $T_nT_{n-1}...T_1$  of transformations  $T_i \in SE_3$  for i = 1, ..., 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, SE<sub>3</sub> is a group and the product is, therefore, some transformation  $T = (R, t) \in SE_3$ . 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  $R \in SO_3$ .

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 Rv = 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 Rv = 1v. 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 SO<sub>3</sub> 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<sup>th</sup> 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 SO<sub>3</sub> 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 SE<sub>2</sub> with the respective rotation matrix in SO<sub>2</sub> 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 SE<sub>2</sub> or SE<sub>3</sub>. Definition 2.1.15 has an advantage over defining the pose of an object as an element of SE<sub>n</sub>: *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 SE<sub>n</sub> would violate the condition of distinguishability since one pose can be represented by more than one element of SE<sub>n</sub>. Nevertheless, working with an element of SE<sub>n</sub> 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  $SE_n$  as in [5, Section 2.2]. For objects without proper symmetry these equivalence classes contain for all poses exactly one element of  $SE_n$  each. If an object is symmetric, the equivalence classes for the object's poses contain more than one element of  $SE_n$ , 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  $T \in SE_n$  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) = Rx + t of the object in the new pose  $P_1$ . We will also denote this as  $T(P_0) = P_1$ .

**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  $T, T' \in SE_n$ , we say T and T' generate the same pose if  $T(P_0) = T'(P_0) \in \mathscr{P}$ . We write

$$T \sim_{P_0} T' \Leftrightarrow T(P_0) = T'(P_0).$$

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

$$\mathcal{T}_{P_0}^{P_1} := \{ T \in SE_n \mid T(P_0) = P_1 \}.$$

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  $SE_n$ , any element of  $SE_n$  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 SO<sub>n</sub> – is more challenging to manage. The mathematical structure of SO<sub>n</sub> 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 SO<sub>2</sub> 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 SO<sub>2</sub> given by the function that maps an  $\alpha \in \mathbb{R}$  to the rotation in SO<sub>2</sub> 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

$$z = \operatorname{Re} z + i \operatorname{Im} z \in \mathbb{C} \iff {\operatorname{Re} z \choose \operatorname{Im} z} \in \mathbb{R}^2.$$

Translating in  $\mathbb{C}$  can be done by adding a number  $z' \in \mathbb{C}$ . 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 SO<sub>2</sub>:

$$\cos \alpha + i \sin \alpha \iff \begin{pmatrix} \cos \alpha & -\sin \alpha \\ \sin \alpha & \cos \alpha \end{pmatrix}.$$

Rotations in  $\mathbb{R}^3$  have various representations, too. The first way of representing a rotation is to describe it with an element of SO<sub>3</sub> as depicted above. According to Theorem 2.1.13, each element of SO<sub>3</sub> 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 SO<sub>3</sub> 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  $R \in SO_3$  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  $R \in SO_3$  that is somehow close to another orientation, on the other hand, cannot be done by adding an arbitrary, "small" matrix  $V \in \mathbb{R}^{3\times3}$  to the matrix R. We know from Theorem 2.1.6 that  $SO_n$  with matrix multiplication is a group. But in general,  $SO_n$  is not closed with respect to matrix addition, as for instance the properties for matrices in  $O_n$  stated in Theorem 2.1.2 are not closed with respect to matrix or vector addition, so R + V will not be in  $SO_3$ , typically. Thus, we have to take a closer look at the geometrical structure of  $SO_3$  [13, Chapter 6]. The special orthogonal group  $SO_n$  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 §17].
  - (ii) Let X and Y be topological spaces. A homeomorphism  $f: X \to Y$  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  $\varphi \colon U \to V$  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  $\mathcal{A} = ((U_i, \varphi_i))_{i \in I}$  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  $(U_1, \varphi_1)$  and  $(U_2, \varphi_2)$  of M with  $U_1 \cap U_2 \neq \emptyset$  is the function

$$\varphi_2 \circ \varphi_1^{-1} \colon \varphi_1(U_1 \cap U_2) \longrightarrow \varphi_2(U_1 \cap U_2).$$

(iv) Two charts  $(U_1, \varphi_1)$  and  $(U_2, \varphi_2)$  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  $(\varphi_2 \circ \varphi_1^{-1})^{-1}$  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  $(U_1, \varphi_1)$ and  $(U_2, \varphi_2)$  are  $C^{\infty}$ -compatible or just compatible. (v) A  $C^k$ -atlas  $\mathcal{A} = ((U_i, \varphi_i))_{i \in I}$  of M is an atlas where the charts  $(U_i, \varphi_i)$  and  $(U_j, \varphi_j)$  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 [1, Section 3.1.5]. Let  $\varphi \colon \mathbb{R}^{m \times n} \to \mathbb{R}^{mn}$  be the function defined by

$$\varphi \begin{pmatrix} a_{11} & a_{12} & \cdots & a_{1n} \\ a_{21} & a_{22} & \cdots & a_{2n} \\ \vdots & \vdots & \ddots & \vdots \\ a_{m1} & a_{m2} & \cdots & a_{mn} \end{pmatrix} = \begin{pmatrix} a_{11} \\ \vdots \\ a_{m1} \\ a_{12} \\ \vdots \\ a_{mn} \end{pmatrix}.$$

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  $\mathbb{R}^{mn}$ . 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 *mn*-dimensional topological manifold, because the topology, the Hausdorff property, and the second-countability are simply transferred via  $\varphi^{-1}$  from  $\mathbb{R}^{mn}$ , and any open neighborhood  $U_A$  of a point  $A \in \mathbb{R}^{m \times n}$ is homeomorphic to the open set  $\varphi(U_A)$  of  $\mathbb{R}^{mn}$ . Furthermore,  $(\mathbb{R}^{m \times n}, \varphi)$  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  $(\mathbb{R}^{m \times n}, \varphi)$ . 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  $(U_1, \varphi_1)$  and  $(U_2, \varphi_2)$  of  $M_1$  and  $M_2$ , the function

$$\psi \colon U_1 \times U_2 \to \mathbb{R}^{d_1} \times \mathbb{R}^{d_2} \colon (x_1, x_2) \mapsto (\varphi_1(x_1), \varphi_2(x_2))$$

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  $\gamma: I \to M$  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  $\gamma \colon I \to M$  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  $(\varphi \circ \gamma_1)'(0) = (\varphi \circ \gamma_2)'(0)$ . The function  $\varphi \circ \gamma_i$  is a mapping  $I \subseteq \mathbb{R} \to \mathbb{R}^n$ , thus  $(\varphi \circ \gamma_i)'(0)$  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*  $T_xM$  to M at x.
- (iv) The tangent bundle

$$TM := \{ (x,\xi) \mid x \in M, \ \xi \in T_x M \}$$

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

$$\vartheta^{x,M}_{\varphi} \colon T_x M \to \mathbb{R}^n \colon \xi = [\gamma] \mapsto (\varphi \circ \gamma)'(0)$$

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  $(\varphi \circ \tilde{\gamma})'(0)$  as  $(\varphi \circ \gamma)'(0)$ . 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  $(\varphi \circ \gamma_1)'(0) \neq (\varphi \circ \gamma_2)'(0)$  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  $\gamma: I \to M$  such that  $(\varphi \circ \gamma)'(0) = p$  and  $\gamma(0) = x$ . Therefore, a reasonable ansatz is given by  $\varphi \circ \gamma(t) = tp + \varphi(x)$ . Indeed, the equivalence class  $\xi$  of the curve

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

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.

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

$$(2.5)$$

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

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

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  $\vartheta_{\psi}^{x,M} \circ (\vartheta_{\varphi}^{x,M})^{-1}$  is linear (see also [37, Section 1.3]). So we can do the following equivalence transformations omitting the indices x and M

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

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 $SO_n$

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

We start with a lemma about the representation of elements of  $SO_n$  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  $R \in SO_n$  can be represented in the form  $R = PBP^T$  with an orthogonal matrix  $P \in O_n$  and a block diagonal matrix

$$B = \operatorname{diag}(R_1(\alpha_1), R_2(\alpha_2), \dots, R_m(\alpha_m), 1, \dots, 1)$$
(2.6)

where  $R_i(\alpha_i) \in SO_2$  for i = 1, ..., m denotes a rotation matrix

$$R_i(\alpha_i) = \begin{pmatrix} \cos \alpha_i & -\sin \alpha_i \\ \sin \alpha_i & \cos \alpha_i \end{pmatrix} \quad \text{with } 0 < \alpha_i \le \pi.$$

*Proof.* Let  $R \in SO_n$  be a rotation matrix. The matrix R is orthogonal, so [21, Theorem 12.10] states that there exist an orthogonal matrix  $P \in O_n$  and a block diagonal matrix  $B = \text{diag}(B_1, B_2, \ldots, B_r)$  where the blocks  $B_j$  are either 1, -1, or of the form

$$B_j = \begin{pmatrix} \cos \theta_j & -\sin \theta_j \\ \sin \theta_j & \cos \theta_j \end{pmatrix} \quad \text{with } 0 < \theta_j < \pi,$$
(2.7)

with  $R = PBP^T$  for j = 1, ..., r. Since  $R \in SO_n$ , the determinant

$$\det(R) = \det(PBP^T) = \det(P)\det(B)\det(P^T) = \det(P)^2\det(B) \stackrel{P \in O_n}{=} \det(B)$$

has to be +1. The determinant of the block diagonal matrix B is given by the product  $\det(B_1)\det(B_2)\cdots\det(B_r)$ . Blocks of the form (2.7) are in SO<sub>2</sub> 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  $\operatorname{diag}(\tilde{B}_1, \tilde{B}_2, \ldots, \tilde{B}_{\tilde{r}})$  with blocks  $\tilde{B}_j$  that are either 1 or of the form

$$\tilde{B}_j = \begin{pmatrix} \cos \tilde{\theta}_j & -\sin \tilde{\theta}_j \\ \sin \tilde{\theta}_j & \cos \tilde{\theta}_j \end{pmatrix} \quad \text{with } 0 < \tilde{\theta}_j \le \pi,$$
(2.8)

for  $j = 1, ..., \tilde{r}$ , 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

$$\underbrace{\begin{pmatrix} 0 & 1 & 0 \\ 0 & 0 & 1 \\ 1 & 0 & 0 \end{pmatrix}}_{=S} \begin{pmatrix} 1 & 0 & 0 \\ 0 & a & b \\ 0 & c & d \end{pmatrix} \underbrace{\begin{pmatrix} 0 & 0 & 1 \\ 1 & 0 & 0 \\ 0 & 1 & 0 \end{pmatrix}}_{=S^T} = \begin{pmatrix} a & b & 0 \\ c & d & 0 \\ 0 & 0 & 1 \end{pmatrix}$$

at the correct positions. After that, B has the form (2.6), and the lemma is proven.  $\Box$ 

In the following, we want to show that  $O_n$  and consequently  $SO_n$  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  $m \ge n$ . Furthermore, let  $F: M \to N$  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  $DF(x)[\cdot]$  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  $O_n \subseteq \mathbb{R}^{n \times n}$ . 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  $O_n$  is a smooth manifold of dimension  $\frac{n(n-1)}{2}$ .

*Proof.* We will show that  $O_n$  is an embedded submanifold of  $\mathbb{R}^{n \times n}$ . Consider the function

$$F: \mathbb{R}^{n \times n} \to \operatorname{Sym}_n: A \mapsto A^T A - I_n.$$

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

Consider the differential DF(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)

$$DF(A)[B] = A^T B + B^T A.$$

The mapping  $DF(A)[\cdot]$  is surjective for every  $A \in O_n$  if for every  $C \in \text{Sym}_n$  there exists a matrix  $B \in \mathbb{R}^{n \times n}$  with DF(A)[B] = C. For  $A \in F^{-1}(\{0_n\}) = O_n$  and  $C \in \text{Sym}_n$ , let  $B = \frac{1}{2}AC$ , resulting in

$$DF(A)[\frac{1}{2}AC] = A^T \frac{1}{2}AC + (\frac{1}{2}AC)^T A = \frac{1}{2}(A^T A C + C^T A^T A) = C$$

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

It follows that  $SO_n$  is a manifold as well if we can show that it is a connected component of  $O_n$ . A connected component of a topological space X is a subset  $C \subseteq X$  that is connected and there exists no larger connected set  $D \supseteq C$  in X [47, Chapter 3 §25].

**Theorem 2.1.28.** The special orthogonal group  $SO_n$  is a smooth manifold of dimension  $\frac{n(n-1)}{2}$ .

*Proof.* We show that  $SO_n$  is a connected component of  $O_n$  as in [61]. Let A be any matrix in  $SO_n$ . The goal is to find an arc in  $SO_n$  that starts from  $I_n$  and ends in A. With Lemma 2.1.25, the matrix A can be represented in the form  $A = PBP^T$  with an orthogonal matrix  $P \in O_n$  and a block diagonal matrix

$$B = \operatorname{diag}(R_1(\alpha_1), R_2(\alpha_2), \dots, R_m(\alpha_m), 1, \dots, 1))$$

for rotations  $\mathbb{R}_i(\alpha_i) \in SO_2$  by the angle  $\alpha_i \in (0, \pi]$ . For a  $\lambda \in [0, 1]$ , let  $B(\lambda)$  define the block diagonal matrix

$$B(\lambda) = \operatorname{diag}(R_1(\lambda \alpha_1), R_2(\lambda \alpha_2), \dots, R_m(\lambda \alpha_m), 1, \dots, 1).$$

and  $A(\lambda) = PB(\lambda)P^T$ . Clearly,  $A(\lambda) \in SO_n$  for all  $\lambda \in [0, 1]$ . Furthermore, this arc starts at  $A(0) = PI_nP^T = I_n$  and ends at B(1) = A. So the special orthogonal group  $SO_n$  is connected. Furthermore, since the function det:  $O_n \to \{-1, 1\}$  is continuous, there cannot be a continuous path in  $O_n$  from a matrix  $C \in O_n$  with det C = 1 to a matrix  $D \in O_n$ with det D = -1. Hence,  $SO_n$  is a connected component of  $O_n$  and therefore open. Thus,  $SO_n$  is an  $\frac{n(n-1)}{2}$ -dimensional topological manifold, since every point  $x \in SO_n$  has

Thus,  $SO_n$  is an  $\frac{n(n-1)}{2}$ -dimensional topological manifold, since every point  $x \in SO_n$  has an open neighborhood  $U_x \subseteq O_n$  in  $O_n$  and therefore an open neighborhood  $U_x \cap SO_n$  in  $SO_n$  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  $O_n$  to  $SO_n$ . Consequently,  $SO_n$  is a smooth manifold of dimension  $\frac{n(n-1)}{2}$ .

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

$$\gamma'(t)\gamma(t)^T + \gamma(t)\gamma'(t)^T = 0_n$$

with the product rule. Since  $\gamma(0) = I_n$ , this reduces to

$$\gamma'(0) + \gamma'(0)^T = 0_n.$$

Thus,  $\gamma'(0)$  is a skew-symmetric matrix. With  $\operatorname{Skew}_n$  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  $T_{I_n} \operatorname{SO}_n$ is a  $\frac{n(n-1)}{2}$ -dimensional vector space over  $\mathbb{R}$  as well. Hence, the spaces  $\operatorname{Skew}_n$  and  $T_{I_n} \operatorname{SO}_n$ are equal (more precisely, they can be identified).

At some arbitrary point  $B \in SO_n$ , let's consider some curve  $\gamma_B: (-1,1) \to SO_n$  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

$$\tilde{\gamma}_B'(0) = B^T \gamma_B'(0) \in T_{I_n} \mathrm{SO}_n = \mathrm{Skew}_n$$

and therefore

$$T_B SO_n = \{ BS \mid S \in Skew_n \}.$$
(2.9)

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  $SE_n$  is a smooth manifold of dimension  $\frac{n(n+1)}{2}$ .

*Proof.* With (2.3) the special Euclidean group can be identified with  $SO_n \times \mathbb{R}^n$ . According to Theorem 2.1.28,  $SE_n$  is the product of two manifolds of dimensions  $\frac{n(n-1)}{2}$  and n. Thus,  $SE_n$  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 SO<sub>3</sub>. Thus, the representation of the poses as elements of SE<sub>3</sub> suffices since a point with an orientation has no symmetry. We can define a simple metric on SO<sub>n</sub> and SE<sub>n</sub> according to [34, Section 3.5].

Lemma 2.1.30. The function

$$d_{\mathrm{SO}_n} \colon \mathrm{SO}_n \times \mathrm{SO}_n \to \mathbb{R}_0^+ \colon (R_1, R_2) \mapsto \|I_3 - R_1 R_2^T\|_F$$

defines a metric on  $SO_n$ , 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

$$||A||_F := \sqrt{\sum_{i,j=1}^n |a_{ij}|^2} = \sqrt{\operatorname{tr}(AA^T)}$$

and according to [58, Section I.3] a norm. Therefore,  $d_{SO_n}(R_1, R_2) = ||I_n - R_1 R_2^T||_F$  is positive or zero. In particular,  $||I_n - R_1 R_2^T||_F = 0$  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  $O_n$ , this is equivalent to  $R_1 = R_2$ . We conclude that  $d_{SO_n}(R_1, R_2) = 0$  if and only if  $R_1 = R_2$ . The computation

$$||I_n - R_1 R_2^T||_F = ||(I_n - R_1 R_2^T)^T||_F = ||I_n - R_2 R_1^T||_F$$

implies that  $d_{SO_n}(R_1, R_2) = d_{SO_n}(R_2, R_1).$ 

To proof the triangle inequality  $d_{SO_n}(R_1, R_2) \leq d_{SO_n}(R_1, R_3) + d_{SO_n}(R_3, R_2)$  for any matrix  $R_3 \in SO_n$ , we start by proving  $||I_n - RS^T||_F = ||S - R||_F$  for all  $R, S \in SO_n$ . With  $||A||_F^2 = \operatorname{tr}(AA^T)$  and  $RR^T = SS^T = I_n$ , we compute

$$||S - R||_F^2 = \operatorname{tr} \left( (S - R)(S - R)^T \right) = \operatorname{tr} \left( SS^T - RS^T - SR^T + RR^T \right) = \operatorname{tr} \left( I_n I_n^T - (RS^T)I_n^T - I_n (RS^T)^T + I_n I_n^T \right) = \operatorname{tr} \left( (I_n - RS^T)(I_n - RS^T)^T \right) = ||I_n - RS^T||_F^2.$$

Thus, we conclude

$$d_{SO_n}(R_1, R_2) = \|I_n - R_1 R_2^T\|_F$$
  
=  $\|R_2 - R_1\|_F$   
=  $\|R_2 - R_3 + R_3 - R_1\|_F$   
 $\leq \|R_2 - R_3\|_F + \|R_3 - R_1\|_F$   
=  $\|I_n - R_3 R_2^T\|_F + \|I_n - R_1 R_3^T\|_H$   
=  $d_{SO_n}(R_1, R_3) + d_{SO_n}(R_3, R_2).$ 

So the triangle inequality holds for  $d_{SO_n}$  and therefore  $d_{SO_n}$  is a metric.

Theorem 2.1.31. The function

$$d_{\mathrm{SE}_n}$$
:  $\mathrm{SE}_n \times \mathrm{SE}_n \to \mathbb{R}_0^+$ :  $(T_1, T_2) = ((R_1, t_1), (R_2, t_2)) \mapsto d_{\mathrm{SO}_n}(R_1, R_2) + d_2(t_1, t_2),$ 

where  $d_2$  is the Euclidean metric on  $\mathbb{R}^n$ , is a metric on  $SE_n$ .

*Proof.* With Lemma 2.1.30,  $d_{SO_n}$  is a metric on  $SO_n$ . Therefore, positivity, symmetry, and the triangle inequality transfer from the metrics  $d_{SO_n}$  and  $d_2$  directly to  $d_{SE_n}$ .

In the following, we will omit the indices  $SO_n$  and  $SE_n$  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 SO<sub>3</sub> manifold by adding an arbitrary small matrix  $V \in \mathbb{R}^{3\times 3}$  to a rotation matrix  $R \in SO_3$ . But if V has a certain structure, it can easily be brought back onto SO<sub>3</sub> 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 matrix  $A \in \mathbb{C}^{n \times n}$  is defined as

$$e^A = \sum_{k\ge 0} \frac{A^k}{k!} \tag{2.10}$$

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  $|a_{ij}|$  of all entries of A. Then for a positive integer  $k \in \mathbb{Z}^+$ , the absolute values of all entries  $a_{ij}^{(k)}$  of the matrix  $A^k$  are bounded by  $(na_{\max})^k$ .

*Proof.* Let  $a_{\max} := \max_{1 \le i,j \le n} |a_{ij}|$  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  $a_{ij}^{(1)} = a_{ij}$ .

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

$$a_{ij}^{(k+1)}| = \left| \sum_{1 \le m \le n} a_{im}^{(k)} a_{mj} \right|$$

$$\stackrel{(1)}{\le} \sum_{1 \le m \le n} |a_{im}^{(k)}| |a_{mj}|$$

$$\stackrel{(2)}{\le} \sum_{1 \le m \le n} (na_{\max})^k a_{\max}$$

$$= n^k \sum_{1 \le m \le n} a_{\max}^{k+1}$$

$$= n^k n a_{\max}^{k+1} = (na_{\max})^{k+1}$$

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

Thus,  $|a_{ij}^{(k+1)}| \leq (na_{\max})^{k+1}$  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  $N \to \infty$ . With the notation and result of Lemma 2.1.33 we can write for the series of one entry

$$\sum_{k \ge 0} \frac{|a_{ij}^{(k)}|}{k!} \le \sum_{k \ge 0} \frac{(na_{\max})^k}{k!} = e^{na_{\max}}$$

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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. BC = CB, then

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

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

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

holds.

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

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

where tr A denotes the trace  $a_{11} + a_{22} + \cdots + a_{nn}$  of A.

- (iv) The exponential of A is regular, i.e.  $e^A \in GL_n$ .
- (v) The inverse of  $e^A$  is given by

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

*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 BC = CB, so

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

This proves the statement.

(ii) Since  $PP^{-1} = I$  and  $(PD^{k-1}P^{-1})(PDP^{-1}) = PD^kP^{-1}$ , induction shows that

$$(PDP^{-1})^k = PD^kP^{-1}$$

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

$$e^{PDP^{-1}} = \sum_{k \ge 0} \frac{(PDP^{-1})^k}{k!} = \sum_{k \ge 0} \frac{PD^kP^{-1}}{k!} = Pe^DP^{-1}.$$

So  $e^{PDP^{-1}} = Pe^{D}P^{-1}$  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  $P \in \operatorname{GL}_n(\mathbb{C})$  with  $A = PJP^{-1}$ .

Part (ii) implies that

$$e^A = e^{PJP^{-1}} = Pe^JP^{-1}.$$

The Jordan matrix J is an upper triangular matrix with the (complex) eigenvalues  $\lambda_i$  for i = 1, ..., n of A in its diagonal (according to their algebraic multiplicity). Again, a simple induction shows that for any upper triangular matrix  $T = (t_{ij})$  the matrix  $e^T$  is also an upper triangular matrix, with the diagonal entries  $e^{t_{ii}}$ . 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

$$\det(e^A) = \det(Pe^JP^{-1}) = \det(P)\det(P)^{-1}\det(e^J) = e^{\lambda_1}e^{\lambda_2}\cdots e^{\lambda_n} = e^{\lambda_1+\lambda_2+\cdots+\lambda_n},$$

where the multiplicativity of the determinant and  $\det(P^{-1}) = \det(P)^{-1}$  was used.

The trace tr A of the matrix A is defined as the sum  $a_{11}+a_{22}+\cdots+a_{nn}$  of the diagonal entries of A. Furthermore, the trace 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 + ib of  $\chi_A$  that lies in  $\mathbb{C} \setminus \mathbb{R}$ , the complex conjugate  $\bar{u} = a - ib$  is a zero of  $\chi_A$ , too. The sum  $u + \bar{u} = 2a$  is in  $\mathbb{R}$ , therefore the sum of all eigenvalues of A is in  $\mathbb{R}$ .

In total, we have

$$\det(e^A) = e^{\lambda_1 + \lambda_2 + \dots + \lambda_n} = e^{\operatorname{tr} A},$$

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

$$e^A e^{-A} = e^{A-A} = e^{0_n} = I_n.$$

**Theorem 2.1.36.** Let  $A \in \mathbb{R}^{2 \times 2}$  be a skew-symmetric matrix of the form

$$A = \begin{pmatrix} 0 & -\theta \\ \theta & 0 \end{pmatrix}$$

Then the exponential of A is given by

matrix of  $e^A$ .

$$e^A = \begin{pmatrix} \cos \theta & -\sin \theta \\ \sin \theta & \cos \theta \end{pmatrix}.$$

Thus,  $e^A$  is in SO<sub>2</sub> [21, Section 18.1].

*Proof.* First, we take a look at the skew-symmetric matrix

$$M := \begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix}.$$

By a simple induction with the induction start

$$M^{2} = \begin{pmatrix} -1 & 0 \\ 0 & -1 \end{pmatrix}, \quad M^{3} = \begin{pmatrix} 0 & 1 \\ -1 & 0 \end{pmatrix}, \quad M^{4} = \begin{pmatrix} 1 & 0 \\ 0 & 1 \end{pmatrix},$$

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

$$A^{4k+1} = \theta^{4k+1} \begin{pmatrix} 0 & -1 \\ 1 & 0 \end{pmatrix} = \theta^{4k+1} M, \quad A^{4k+2} = \theta^{4k+2} \begin{pmatrix} -1 & 0 \\ 0 & -1 \end{pmatrix} = -\theta^{4k+2} I_{2},$$
$$A^{4k+3} = \theta^{4k+3} \begin{pmatrix} 0 & 1 \\ -1 & 0 \end{pmatrix} = -\theta^{4k+3} M, \quad A^{4k+4} = \theta^{4k+4} \begin{pmatrix} 1 & 0 \\ 0 & 1 \end{pmatrix} = \theta^{4k+4} I_{2}.$$

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

$$e^{A} = \sum_{k \ge 0} \frac{A^{k}}{k!} = I_{3} + \frac{A}{1!} + \frac{A^{2}}{2!} + \frac{A^{3}}{3!} + \frac{A^{4}}{4!} + \frac{A^{5}}{5!} + \dots$$

$$= \left(I_{3} + \frac{1}{2!}A^{2} + \frac{1}{4!}A^{4} + \dots\right) + \left(A + \frac{1}{3!}A^{3} + \frac{1}{5!}A^{5} + \dots\right)$$

$$= \left(\theta^{0}I_{2} + \frac{1}{2!}(-\theta^{2}I_{2}) + \frac{1}{4!}\theta^{4}I_{2} + \dots\right) + \left(\theta M + \frac{1}{3!}(-\theta^{3}M) + \frac{1}{5!}\theta^{5}M + \dots\right)$$

$$= \left(\sum_{k \ge 0} (-1)^{k} \frac{\theta^{2k}}{(2k)!}\right) I_{2} + \left(\sum_{k \ge 0} (-1)^{k} \frac{\theta^{2k+1}}{(2k+1)!}\right) M$$

$$= \cos \theta I_{2} + \sin \theta M = \left( \begin{array}{c} \cos \theta & -\sin \theta \\ \sin \theta & \cos \theta \end{array} \right).$$

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 SO<sub>2</sub>. This even holds for skew-symmetric matrices in  $\mathbb{R}^{n \times n}$  for arbitrary  $n \ge 2$ . Gallier gives in [21, Theorem 18.1] the following statement.

**Theorem 2.1.37.** Let  $\operatorname{Skew}_n \subseteq \mathbb{R}^{n \times n}$  be the set of all real skew-symmetric  $(n \times n)$ -matrices.

- (i) For all  $A \in \text{Skew}_n$ , the exponential  $e^A$  is an element of  $SO_n$ .
- (ii) The exponential map

$$e^{\cdot} : \operatorname{Skew}_n \to \operatorname{SO}_n$$
 (2.11)

with the domain  $Skew_n$  and the codomain  $SO_n$  is surjective.

*Proof.* (i) Let  $A \in \text{Skew}_n$  be a skew-symmetric matrix. First, we show that  $A \in O_n$ . Since A is skew-symmetric, we know  $A^T = -A$ . Considering the power series representation of  $e^A$ , we see that  $(e^A)^T = e^{A^T}$ . With Lemma 2.1.35(v) follows

$$(e^{A})^{T} e^{A} = e^{A^{T}} e^{A} = e^{-A} e^{A} = I_{T}$$

and  $e^A(e^A)^T = I_n$  analogously. Thus, the matrix  $e^A$  is orthogonal, i.e.

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

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

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

To sum up, from  $e^A \in \mathcal{O}_n$  and  $\det(e^A) = 1$  follows that  $e^A \in \mathcal{SO}_n$ .

(ii) Part (i) guarantees that the function (2.11) is well-defined. Let  $R \in SO_n$  be a rotation matrix. With Lemma 2.1.25, the matrix R can be represented in the form  $R = PBP^T$  with an orthogonal matrix  $P \in O_n$  and a block diagonal matrix

$$B = \operatorname{diag}(B_1, B_2, \dots, B_r, 1, \dots, 1).$$

The blocks  $B_j \in SO_2$  are of the form

$$B_j = \begin{pmatrix} \cos \theta_j & -\sin \theta_j \\ \sin \theta_j & \cos \theta_j \end{pmatrix} \quad \text{with } 0 < \theta_j \le \pi.$$

We have to find a skew-symmetric matrix A with  $e^A = R$ . Let's define a block diagonal matrix  $C = \text{diag}(C_1, C_2, \ldots, C_r, 0, \ldots, 0)$  as follows. For a block  $B_j$  let  $C_j$  be the  $(2 \times 2)$ -matrix

$$C_j := \begin{pmatrix} 0 & -\theta_j \\ \theta_j & 0 \end{pmatrix}.$$
(2.12)

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

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

The exponential  $e^{C_j}$  for j = 1, ..., 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  $PCP^T$  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

$$A + A^{T} = PCP^{T} + (PCP^{T})^{T} = PCP^{T} + PC^{T}P^{T} = P(C + C^{T})P^{T} = 0_{n}$$

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

$$e^A = e^{PCP^T} \stackrel{2.1.35(ii)}{=} Pe^C P^T = PBP^T = R$$

with the considerations above and Lemma 2.1.35(ii).

So, for an arbitrary rotation matrix  $R \in SO_n$ , we found a skew-symmetric matrix  $A \in Skew_n$  such that  $e^A = R$ . Thus, the exponential map is surjective with domain Skew<sub>n</sub> and codomain SO<sub>n</sub>.

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

$$A = \begin{pmatrix} 0 & -c & b \\ c & 0 & -a \\ -b & a & 0 \end{pmatrix}$$

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

$$\tilde{A} := \begin{pmatrix} a^2 & ab & ac \\ ab & b^2 & bc \\ ac & bc & c^2 \end{pmatrix}.$$

The computation

$$A\tilde{A} = \begin{pmatrix} 0 & -c & b \\ c & 0 & -a \\ -b & a & 0 \end{pmatrix} \begin{pmatrix} a^2 & ab & ac \\ ab & b^2 & bc \\ ac & bc & c^2 \end{pmatrix} = \begin{pmatrix} -cab + bac & -cb^2 + b^2c & -cbc + bc^2 \\ ca^2 - a^2c & cab - abc & cac - ac^2 \\ -ba^2 + a^2b & -bab + ab^2 & -bac + abc \end{pmatrix} = 0_3$$

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

$$A^{2} = \begin{pmatrix} -c^{2} - b^{2} & ba & ca \\ ab & -c^{2} - a^{2} & cb \\ ac & bc & -b^{2} - a^{2} \end{pmatrix}$$
$$= \begin{pmatrix} -c^{2} - b^{2} - a^{2} + a^{2} & ab & ac \\ ab & -c^{2} - a^{2} - b^{2} + b^{2} & bc \\ ac & bc & -b^{2} - a^{2} - c^{2} + c^{2} \end{pmatrix} = \tilde{A} - \theta^{2}I_{3}$$

Multiplying this equation by A gives

$$A^3 = A(\tilde{A} - \theta^2 I_3) = -\theta^2 A$$

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

$$A^{4k+1} = \theta^{4k}A \qquad |\cdot A$$

$$A^{4k+2} = \theta^{4k} A^2 \qquad \qquad | \cdot A$$

$$A^{4k+3} = \theta^{4k} A^3 = \theta^{4k} (-\theta^2 A) = -\theta^{4k+2} A \qquad | \cdot A$$
$$A^{4k+4} = -\theta^{4k+2} A^2$$

by assuming  $A^{4k} = -\theta^{4k-2}A^2$  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

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

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  $g: D \to R$  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  $f_j: D_j \to R$  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

$$g(x_1, \dots, x_n) = \prod_{j \in J} f_j(X_j)$$
 (2.13)

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with  $X_j = (x_{j_1}, \ldots, x_{j_k})$  where  $f_j(X_j)$  stands for  $f_j(x_{j_1}, \ldots, x_{j_k})$ . A factor graph for this factorization of g is a graph  $\mathcal{F} = (V, E)$  with the set of vertices  $V = V_v \cup V_f$  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  $g(x_1, \ldots, x_n)$ , 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  $\{f_j, x_i\} \in E$  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  $\mathcal{F} = (V_v, V_f, E)$  for the factor graph  $\mathcal{F}$ .



Figure 2.2.: Factor graph for the product  $f_a(x_1)f_b(x_2, x_4)f_c(x_1, x_2, x_5)f_d(x_3, x_4, x_5)$ .

*Example 2.2.3.* Let  $g: \mathbb{Z}_3^5 \to \mathbb{Z}_3$  be a function with the factorization

$$g(x_1, x_2, x_3, x_4, x_5) = \underbrace{x_1^2}_{=:f_a(x_1)} \underbrace{(2x_2 + x_4 + 1)}_{=:f_b(x_2, x_4)} \underbrace{(x_1 + x_2^4 + x_5^3 + 2)}_{=:f_c(x_1, x_2, x_5)} \underbrace{(x_3 + x_4 + x_5)}_{=:f_d(x_3, x_4, x_5)}.$$

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

$$g(x_1, x_2, x_3, x_4, x_5) = \underbrace{x_1^2(2x_2 + x_4 + 1)}_{=:f_e(x_1, x_2, x_4)} \underbrace{(x_1 + x_2^4 + x_5^3 + 2)}_{=:f_e(x_1, x_2, x_5)} \underbrace{(x_3 + x_4 + x_5)}_{=:f_d(x_3, x_4, x_5)},$$

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  $G = (V_1, V_2, E)$  of vertices  $V = V_1 \cup V_2$  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  $g: D \to R$  with a factorization  $\prod_{j \in J} f_j(X_j)$  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  $g(x_1, \ldots, x_n)$  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|Z) [13, Section 1.6]. The following lemma can rephrase this maximization problem.

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

$$X^{\text{MAP}} = \operatorname*{arg\,max}_{X} p(X, Z)$$

*Proof.* Bayes' law states in this context that

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

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|Z) translates as

$$\underset{X}{\operatorname{arg\,max}} p(X|Z) = \underset{X}{\operatorname{arg\,max}} \frac{p(Z|X)p(X)}{p(Z)} = \underset{X}{\operatorname{arg\,max}} p(Z|X)p(X).$$

The term p(Z|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  $X^{MAP}$ can be computed by maximizing the joint probability p(X, Z). 

Remark 2.2.6. The term p(Z|X) is also called likelihood (function) in statistics and can be denoted with L(X|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 Vof the factor graph  $\mathcal{F} = (V, F, E)$  is given by  $V = \{x_1, x_2, x_3, l_1, l_2, l_3\}$  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  $p(z_{l_j}^{(i)}|x_i, l_j)$ . With the definition of conditional probability [6, Section 1.3], the joint probability

 $p(x_1, x_2, x_3)$  is given by

$$p(x_1, x_2, x_3) = p(x_3 | x_2, x_1) p(x_2 | x_1) p(x_1) = p(x_3 | x_2) p(x_2 | x_1) p(x_1)$$

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

$$p(X,Z) = p(x_3|x_2)p(x_2|x_1)p(x_1)p(l_1)p(l_2)p(l_3)\prod_{i,j}p(z_{l_j}^{(i)}|x_i,l_j),$$

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  $(x_1, x_2, \text{ and } x_3)$  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  $I_C: A^n \to \{0, 1\}$  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  $g(x) = x^2 - 3x + 2$  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 SO<sub>2</sub> and SO<sub>3</sub> in Section 2.3.3. Section 2.3.4 deals with optimization in SE<sub>3</sub>. 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 GT-SAM [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

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

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  $-\operatorname{grad} g(x^{(t)})$  of the current guess  $x^{(t)}$ . Thus, the update rule for gradient descent is given by

$$x^{(t+1)} = x^{(t)} + \alpha \delta_{\text{GD}}^{(t)} = x^{(t)} - \alpha \text{ grad } g(x^{(t)}).$$

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

$$\min_{x \in \mathbb{R}^n} \sum_{i=1}^m (g_i(x))^2$$
(2.15)

with the function  $g = (g_1, \ldots, g_m)$  and  $g_i \colon \mathbb{R}^n \to \mathbb{R}$  for each  $i = 1, \ldots, m$ . The idea is to approximate g by a Taylor series of first order [45], thus

$$g(x) \approx g(x_0) + \sum_{j=1}^{n} \operatorname{grad}_j g(x_0)(x - x_0)_j = g(x_0) + J(x_0)(x - x_0)$$
(2.16)

where J denotes the well-known Jacobian matrix

$$J(x) := \begin{pmatrix} \frac{\partial g_1}{\partial x_1}(x) & \cdots & \frac{\partial g_1}{\partial x_n}(x) \\ \vdots & \ddots & \vdots \\ \frac{\partial g_m}{\partial x_1}(x) & \cdots & \frac{\partial g_m}{\partial x_n}(x) \end{pmatrix}.$$

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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

grad 
$$||g(x_0) + J(x_0)(x - x_0)||^2 = 0 \Rightarrow J^T(x_0) (J(x_0)(x - x_0) + g(x_0)) = 0.$$

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

$$J^{T}(x^{(t)})J(x^{(t)})\delta^{(t)}_{\rm GN} = -J^{T}(x^{(t)})g(x^{(t)}).$$
(2.17)

Thus, the Gauss-Newton update rule is

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

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

$$\left(J^T(x^{(t)})J(x^{(t)}) + \lambda \operatorname{diag}\left(J^T(x^{(t)})J(x^{(t)})\right)\right)\delta^{(t)}_{\mathrm{LM}} = -J^T(x^{(t)})g(x^{(t)}).$$
(2.18)

where diag(A) denotes the diagonal matrix diag( $a_{11}, a_{22}, \ldots, a_{ll}$ ) 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

 $\mathcal{R}\colon TM\to M$ 

such that the following properties hold for the restriction  $\mathcal{R}_x := \mathcal{R}|_{T_xM}$  for every  $x \in M$ (i.e. in  $\mathcal{R}_x$ , we drop the first part of a pair  $(x,\xi) \in TM$ ).

(i) Let  $0_x$  denote the zero vector of the vector space  $T_x M$ , then  $\mathcal{R}_x(0_x) = x$ .

(ii) The differential  $D\mathcal{R}_x(0_x)[\cdot]$  is equal to the identity  $\mathrm{id}_{T_xM}$  on the tangent space  $T_xM$ .

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  $(\varphi \circ \gamma)'(0) = 0_n$ .

Part (ii) of Definition 2.3.1 is often referred to as the *local rigidity condition* [1, Section 4.1]. The differential  $D\mathcal{R}_x(0_x)[\cdot]$  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  $D\mathcal{R}_x(0_x)[\eta]$  in the tangent space of the codomain of  $\mathcal{R}_x$ . So to be precise,  $D\mathcal{R}_x(0_x)[\cdot]$  is a function  $T_{0_x}(T_xM) \to T_xM$  and is only equal to the identity on  $T_xM$  if we identify  $T_{0_x}(T_xM)$  and  $T_xM$  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  $c: M \to \mathbb{R}$  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

$$c^{\mathcal{R}} := c \circ \mathcal{R} \colon TM \to \mathbb{R}$$

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

$$c_x^{\mathcal{R}} := c^{\mathcal{R}}|_{T_xM} = c \circ \mathcal{R}_x \colon T_xM \to \mathbb{R}$$

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  $Dc_x^{\mathcal{R}}(0_x) = Dc(x)$ .

Remark 2.3.3. Absil, Mahony, and Sepulchre state in [1, (4.4)] that even

$$\operatorname{grad} c_x^{\mathcal{R}}(0_x) = \operatorname{grad} c(x)$$

holds if the manifold M is endowed with a *Riemannian metric* (which is the case for  $SO_n$ ). 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  $SO_n$ . We will focus on a retraction that uses the exponential map according to [2, Example 2]. For a rotation  $R \in SO_n$  a retraction is given by

$$\mathcal{R}_R: T_R \mathrm{SO}_n \to \mathrm{SO}_n: \eta \mapsto Re^{R^{-1}\eta}.$$
(2.19)

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

$$\mathcal{R}_R(\eta) = Re^{R^{-1}\eta} = Re^{R^{-1}R\hat{\xi}} = Re^{\hat{\xi}}.$$

The exponential  $e^{\hat{\xi}}$  of the skew-symmetric matrix  $\hat{\xi}$  is according to Theorem 2.1.37 in SO<sub>n</sub>. Thus, the matrix product  $Re^{\hat{\xi}}$  is in SO<sub>n</sub> 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  $(R, \eta) \in TSO_n$  to  $\mathcal{R}_R(\eta)$ . Condition (i) holds due to the computation

$$\mathcal{R}_{R}(0_{R}) = Re^{R^{-1}0_{R}} = Re^{R^{-1}R0_{n}} = Re^{0_{n}} = RI_{n} = R$$

with the zero element  $0_R$  of the tangent space  $T_R SO_n$  and the zero element  $0_n$  of the vector space Skew<sub>n</sub>. For the second part of the definition, we have to compute the differential  $D\mathcal{R}(0_R)[\cdot]$ . Najfeld and Havel give in [48, Section 1.2] the general definition

$$De^{tA}[V] = \lim_{h \to 0} \frac{1}{h} \left( e^{t(A+hV)} - e^{tA} \right)$$

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

$$De^{tA}[V] = \int_0^t e^{(t-\tau)A} V e^{\tau A} d\tau.$$

Here, with  $\eta = R\hat{\xi}_{\eta} \in T_R SO_n$  and  $\zeta = R\hat{\xi}_{\zeta} \in T_R SO_n$ , this translates to

$$D\mathcal{R}_{R}(0_{R})[\zeta] = \lim_{h \to 0} \frac{1}{h} \left( Re^{R^{-1}(\eta + h\zeta)} - Re^{R^{-1}\eta} \right) \Big|_{\eta = 0_{R}}$$
  
=  $\lim_{h \to 0} \frac{1}{h} \left( Re^{R^{-1}(R\hat{\xi}_{\eta} + hR\hat{\xi}_{\zeta})} - Re^{R^{-1}R\hat{\xi}_{\eta}} \right) \Big|_{R\hat{\xi}_{\eta} = 0_{n}}$   
=  $R \lim_{h \to 0} \frac{1}{h} \left( e^{\hat{\xi}_{\eta} + h\hat{\xi}_{\zeta}} - e^{\hat{\xi}_{\eta}} \right) \Big|_{\hat{\xi}_{\eta} = 0_{n}}$   
=  $R De^{0_{n}}[\hat{\xi}_{\zeta}]$   
=  $R \int_{0}^{1} e^{(1-\tau)0_{n}}\hat{\xi}_{\zeta}e^{\tau 0_{n}}d\tau$   
=  $R \int_{0}^{1} I_{n}\hat{\xi}_{\zeta}I_{n}d\tau = R\hat{\xi}_{\zeta} = \zeta.$ 

This shows that  $D\mathcal{R}_R(0_R)[\cdot]$  is the identity on  $T_RSO_n$ . Thus,  $\mathcal{R}$  is a retraction.

### **2.3.3.** Optimization on $SO_2$ and $SO_3$

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 SO<sub>2</sub>, this process is quite straightforward as described in [13, Section 6.1.5]. The special orthogonal group SO<sub>2</sub> is according to Theorem 2.1.28 a  $\left(\frac{2(2-1)}{2} = 1\right)$ -dimensional manifold. The tangent space  $T_B$ SO<sub>2</sub> at some point  $B \in$  SO<sub>2</sub>, given by (2.9), is a 1-dimensional 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 SO<sub>3</sub>-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 SO<sub>2</sub> and SO<sub>3</sub>.

of optimizing on the rotation manifolds SO<sub>2</sub> and SO<sub>3</sub>. The special orthogonal group SO<sub>3</sub> is a  $(\frac{3(3-1)}{2} = 3)$ -dimensional manifold with the tangent space  $T_B$ SO<sub>3</sub> = { $BS \mid S \in$ Skew<sub>3</sub>}. 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 SO<sub>2</sub> and SO<sub>3</sub>, 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<sub>3</sub> 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  $c = (c_1, c_2)^T$  to its orthogonal vector  $(-c_2, c_1)^T$ . 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

$$\hat{}: \quad \mathbb{R} \to \mathbb{R}^{2 \times 2} \colon \quad \xi \mapsto \hat{\xi} := \begin{pmatrix} 0 & -\xi \\ \xi & 0 \end{pmatrix}.$$

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

$$: \mathbb{R}^3 \to \mathbb{R}^{3 \times 3} : \xi = \begin{pmatrix} \xi_x \\ \xi_y \\ \xi_z \end{pmatrix} \mapsto \hat{\xi} := \begin{pmatrix} 0 & -\xi_z & \xi_y \\ \xi_z & 0 & -\xi_x \\ -\xi_y & \xi_x & 0 \end{pmatrix}.$$



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 §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  $T_B SO_3$  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  $T_B SO_3$  given by the product of B with  $\hat{\xi}$ .

So, when optimizing in SO<sub>2</sub>, the hat-operator transforms the rotation angle  $\xi$  into a skew-symmetric (2 × 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

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

$$\stackrel{2.1.36}{=} \begin{pmatrix} \cos\theta & -\sin\theta\\ \sin\theta & \cos\theta \end{pmatrix} \begin{pmatrix} \cos\xi & -\sin\xi\\ \sin\xi & \cos\xi \end{pmatrix}$$

$$\stackrel{2.1.9}{=} \begin{pmatrix} \cos(\theta+\xi) & -\sin(\theta+\xi)\\ \sin(\theta+\xi) & \cos(\theta+\xi) \end{pmatrix}.$$

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 SO<sub>2</sub> or SO<sub>3</sub> 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

$$R_0 \oplus \xi := \mathcal{R}_{R_0}(R_0\hat{\xi})$$

Note, that the matrix  $e^{\xi}$  for  $\xi = \alpha a$ , that appears in a local update in SO<sub>3</sub>, 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

$$\|\xi\| = \sqrt{(\alpha a_x)^2 + (\alpha a_y)^2 + (\alpha a_z)^2} = |\alpha| \|a\| = |\alpha|$$

holds. Hence, this retraction can be computed efficiently in  $SO_2$  and  $SO_3$ .

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 SO<sub>3</sub>. 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  $z = (z^{(e)}, z^{(w_1)}, z^{(w_2)}, z^{(w_3)})^T \in \mathbb{R}^{12}$  where  $z^{(e)}, z^{(w_1)}, z^{(w_2)}, z^{(w_3)} \in \mathbb{R}^3$  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

 $h: \mathrm{SO}_3 \to \mathbb{R}^{12}.$ 

For a given rotation  $R \in SO_3$ , 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  $h(SO_3)$  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  $h(SO_3)$ . 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

$$R^* = \underset{R \in SO_3}{\arg\min} \|h(R) - z\|^2.$$

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  $h: SO_3 \to \mathbb{R}^n$ , the goal is to find

$$R^* = \underset{R \in SO_3}{\arg\min} \|h(R) - z\|^2$$

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  $t \ge 0$  by minimizing

$$\delta^{(t)} = \operatorname*{arg\,min}_{\xi \in \mathbb{R}^3} \|h(R^{(t)} \oplus \xi) - z\|^2$$

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  $\delta_{\text{LM}}^{(t)}$  as in (2.18).

## **2.3.4.** Optimization in $SE_3$

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  $SE_3$  rather than just optimal rotations. This is just a simple generalization from  $SO_3$  to  $SE_3$ , analogously to [13, Sections 6.2.2 and 6.2.3].

Local coordinates  $\xi$  in SO<sub>3</sub> were given by the product of an axis  $a \in S^2$  and an angle  $\alpha \in \mathbb{R}$ . For local coordinates in SE<sub>3</sub>, we simply expand the local coordinates of SO<sub>3</sub> 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 SE<sub>3</sub> (see Theorem 2.1.29).

The tangent space  $T_{(R,t)}SE_3$  of the product manifold  $SE_3 = SO_3 \times \mathbb{R}^3$  is according to [59, solution to Problem 8.7] isomorphic to the product of the tangent spaces  $T_RSO_3 \times T_t\mathbb{R}^3$ . Thus, we view tangent vectors to  $SE_3$  as a pair  $(\eta, s) \in T_RSO_3 \times T_t\mathbb{R}^3$  or a matrix

$$\begin{pmatrix} \eta & s \\ 0 & 0 \end{pmatrix} \in \mathbb{R}^{4 \times 4}$$

if necessary, with  $\eta \in RSkew_3$  according to (2.9) and  $s \in \mathbb{R}^3$ .

Therefore, we can define a retraction for the special Euclidean group SE<sub>3</sub> as in Example 2.3.4. For  $(R, t) \in SE_3$  and  $(\eta, s) = (R\hat{\xi}_{SO_3}, s) \in T_RSO_3 \times \mathbb{R}^3$ , we define the retraction as

$$\mathcal{R}_{(R,t)} \colon T_{(R,t)} \mathrm{SE}_3 \to \mathrm{SE}_3 \colon (\eta, s) \mapsto \begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \begin{pmatrix} e^{R^{-1}\eta} & R^{-1}s \\ 0 & 1 \end{pmatrix} = \begin{pmatrix} Re^{R^{-1}\eta} & s+t \\ 0 & 1 \end{pmatrix}.$$
(2.20)

We see that  $\mathcal{R}_{(R,t)}(\eta, s) = (\mathcal{R}_R^{SO_3}(\eta), s+t) \in SE_3$  with the retraction  $\mathcal{R}_R^{SO_3}$  of Example 2.3.4. At the zero element  $0_{(R,t)}$  of  $T_{(R,t)}SE_3$ , we compute

$$\mathcal{R}_{(R,t)}(0_{(R,t)}) = \begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \begin{pmatrix} e^{R^{-1}0_R} & R^{-1}0 \\ 0 & 1 \end{pmatrix} = \begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} I_4 = (R,t).$$

With the computation of the differential  $D\mathcal{R}_R^{SO_3}(0_R)[\zeta]$  in Example 2.3.4 we conclude for some tangent vector  $(\zeta, u) = (R\hat{\xi}_{\zeta}, u) \in T_{(R,t)}SE_3$ 

$$D\mathcal{R}_{(R,t)}(0_{(R,t)})[(\zeta, u)] = \lim_{h \to 0} \frac{1}{h} \left( \mathcal{R}_{(R,t)}(\eta + h\zeta, s + hu) - \mathcal{R}_{(R,t)}(\eta, s) \right) \Big|_{(\eta,s) = (0_R, 0_3)}$$
  
=  $\begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \begin{pmatrix} \hat{\xi}_{\zeta} & \lim_{h \to 0} \frac{1}{h} R^{-1}hu \\ 0 & \lim_{h \to 0} \frac{1}{h} \end{pmatrix}$   
=  $\begin{pmatrix} R & t \\ 0 & 1 \end{pmatrix} \begin{pmatrix} \hat{\xi}_{\zeta} & R^{-1}u \\ 0 & 0 \end{pmatrix}$   
=  $\begin{pmatrix} R\hat{\xi}_{\zeta} & RR^{-1}u + 0t \\ 0 & 0 \end{pmatrix}$   
=  $(\zeta, u).$ 

This proves that the function defined in (2.20) is indeed a retraction for SE<sub>3</sub> in the sense of Definition 2.3.1. We can use this retraction to define local updates in SE<sub>3</sub>.

**Definition 2.3.8.** For  $T_0 = (R_0, t_0) \in SE_3$  and local coordinates

$$\xi = \begin{pmatrix} \omega \\ v \end{pmatrix} \in \mathbb{R}^6$$

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  $h: SE_3 \to \mathbb{R}^n$  be an estimation function. To find

$$T^* = \underset{T \in SE_2}{\arg\min} \|h(T) - z\|^2$$
(2.21)

for some measurement z, we compute the step

$$\delta^{(t)} = \operatorname*{arg\,min}_{\xi \in \mathbb{R}^6} \|h(T^{(t)} \oplus \xi) - z\|^2$$

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

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## 2.3.5. Optimizing a Factor Graph

Now that we know how to optimize for poses in  $SE_3$ , 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  $SE_3$  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  $f(X) = \prod_j f_j(X_j)$  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  $X^{\max}$  maximizes the global function f, i.e.

$$X^{\max} = \underset{X}{\arg\max} f(X) = \underset{X}{\arg\max} \prod_{j} f_j(X_j).$$

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|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  $f_j(X_j)$  are probability densities of some multivariate Gaussian distribution. As a reminder, the multivariate Gaussian distribution  $\mathcal{N}(\mu, \Sigma)$  is given by the density

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

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

$$f_j(X_j) \propto e^{-\frac{1}{2} \left( h(X_j) - z_j \right)^T \sum_j^{-1} \left( h(X_j) - z_j \right) \right)} = e^{-\frac{1}{2} \| h(X_j) - z_j \|_{\Sigma_j}^2}$$
(2.22)

with the notation  $(h(X_j) - z_j)^T \Sigma_j^{-1} (h(X_j) - z_j) = ||h(X_j) - z_j)||_{\Sigma_j}^2$  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  $h(X_j) - z_j$  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

X

$$\begin{array}{l} \text{MAP} \stackrel{2.2.5}{=} \arg\max_{X} p(X,Z) \\ = & \arg\max_{X} \prod_{j} f_{j}(X_{j}) \\ \stackrel{(1)}{=} & \arg\max_{X} \prod_{j} e^{-\frac{1}{2} \|h_{j}(X_{j}) - z_{j}\|_{\Sigma_{j}}^{2}} \\ \stackrel{(2)}{=} & \arg\max_{X} \log \left( \prod_{j} e^{-\frac{1}{2} \|h_{j}(X_{j}) - z_{j}\|_{\Sigma_{j}}^{2}} \right) \\ = & \arg\max_{X} - \frac{1}{2} \sum_{j} \|h_{j}(X_{j}) - z_{j}\|_{\Sigma_{j}}^{2} \\ \stackrel{(3)}{=} & \arg\min_{X} \sum_{j} \|h_{j}(X_{j}) - z_{j}\|_{\Sigma_{j}}^{2}. \end{array}$$

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  $X_i$  in some manifold, the resulting optimization problem

$$X^{\text{MAP}} = \arg\min_{X} \sum_{j} \|h_{j}(X_{j}) - z_{j}\|_{\Sigma_{j}}^{2}$$
(2.23)

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  $SE_3$  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].







(c) Downsampled point cloud with x = 0.018m. (d) Downsampled point cloud with x = 0.036m.

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 C that lie within a certain distance from a reference point pnaively, we just iterate over all points in C, compute the distance, and check if this distance is below a certain threshold. Computing the neighbors of all n points in C takes in total  $\Theta(n^2)$  time, which is pretty bad. Thus, we use and briefly describe kd-trees, introduced by Bentley in [4].

This is a data structure for k-dimensional data. Here, we deal with 3-dimensional points  $p = (p_1, p_2, p_3)^T \in \mathbb{R}^3$ , 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 C into two sets  $C_1, C_2$  for the two sides of the plane. The set  $C_1$  is forwarded to the left child of the root and the set  $C_2$  to the right child. We split  $C_1$  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  $C_{11}$  and  $C_{12}$ . Then,

we split  $C_{11}$  according to their third coordinate (we divide the space with a plane parallel to the first and second axis) into the sets  $C_{111}$  and  $C_{112}$ . After that, we start again by dividing the points according to their first coordinate. We continue for each set  $C_{n_1n_2...n_l}$ 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 kd-tree with an axis-aligned search region takes  $O(n^{1-\frac{1}{k}} + m)$  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 kd-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 kd-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 kd-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  $f(x) = ax^2 + bx + c$  in a set of data points  $\{(x_i, y_i)_{i=1}^N\} \subseteq \mathbb{R}^2$  using least-squares regression is the task of minimizing the sum of squared errors

$$\sum_{i=1}^{N} (y_i - f(x_i))^2$$

with respect to the parameters  $a, b, c \in \mathbb{R}$  [49].

Fitting a plane given of the equation E: ax + by + cz + d = 0 in a set of data points  $\{(x_i, y_i, z_i)_{i=1}^N\} \subseteq \mathbb{R}^3$  using orthogonal regression is the task of minimizing the sum of squared orthogonal distances

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

with respect to the parameters  $a, b, c, d \in \mathbb{R}$ , where  $d(E, p_i)$  denotes the orthogonal distance of the point  $p_i = (x_i, y_i, z_i)$  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

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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 mpoints. 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 C, model tolerance  $\varepsilon$ , maximum number of iterations N **Output:** best model parameters M

 $\begin{array}{l} M_{\mathrm{best}} \leftarrow \mathrm{None} \\ n_{\mathrm{best}} \leftarrow 0 \\ \mathbf{for} \ i \leftarrow 1 \ \mathbf{to} \ N \ \mathbf{do} \\ \mathrm{randomly \ sample} \ m \ \mathrm{points} \ \mathrm{from} \ \mathcal{C} \\ M \leftarrow \mathrm{parameters} \ \mathrm{for} \ \mathrm{the} \ \mathrm{model} \ \mathrm{defined} \ \mathrm{by} \ \mathrm{the} \ \mathrm{sampled} \ \mathrm{points} \\ n \leftarrow \mathrm{number} \ \mathrm{of} \ \mathrm{points} \ \mathrm{in} \ \mathcal{C} \ \mathrm{with} \ \mathrm{a} \ \mathrm{distance} < \varepsilon \ \mathrm{to} \ \mathrm{the} \ \mathrm{model} \ \mathrm{with} \ \mathrm{parameters} \ M \\ \mathbf{if} \ n > n_{\mathrm{best}} \ \mathbf{then} \\ n_{\mathrm{best}} \leftarrow n \\ M_{\mathrm{best}} \leftarrow M \\ \mathbf{end} \ \mathbf{if} \\ \mathbf{end} \ \mathbf{for} \\ \mathbf{return} \ M_{\mathrm{best}} \end{array}$ 

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  $\Theta(N^2)$ . 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 C of c points and the true optimal solution  $S \subseteq C$  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 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  $\mathbb{E}(K_m)$  of the number of trials is given by

$$\mathbb{E}(K_m) = \sum_{k \ge 1} \mathbb{P}(K_m = k)k,$$

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

$$q_m := \frac{\binom{n}{m}}{\binom{c}{m}} = \frac{n!}{m!(n-m)!} \frac{m!(c-m)!}{c!} = \frac{n!(c-m)!}{c!(n-m)!}$$
(2.24)

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

$$\mathbb{P}(K_m = k) = (1 - q_m)^{k-1} q_m.$$

Now we can compute the expected value as

$$\mathbb{E}(K_m) = \sum_{k \ge 1} (1 - q_m)^{k-1} q_m k$$

$$\stackrel{(1)}{=} q_m \sum_{k \ge 1} k \bar{q}_m^{k-1}$$

$$\stackrel{(2)}{=} q_m \left( \sum_{k \ge 0} \bar{q}_m^k \right)'$$

$$\stackrel{(3)}{=} q_m \left( \frac{1}{1 - \bar{q}_m} \right)'$$

$$= q_m (-1) \frac{-1}{(1 - \bar{q}_m)^2} = \frac{q_m}{q_m^2} = \frac{1}{q_m}.$$

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  $\mathbb{V}(K_m)$  of the number of iterations till success is then given by

$$\begin{aligned} \mathbb{V}(K_m) &= \mathbb{E}(K_m^2) - \mathbb{E}(K_m)^2 \\ &= \sum_{k \ge 1} \mathbb{P}(K_m = k)k^2 - \left(\sum_{k \ge 1} \mathbb{P}(K_m = k)k\right)^2 \\ &= \sum_{k \ge 1} (1 - q_m)^{k-1} q_m k^2 - \left(\frac{1}{q_m}\right)^2 \\ &= q_m \sum_{k \ge 1} k(k - 1 + 1)\bar{q}_m^{k-1} - \frac{1}{q_m^2} \\ &\stackrel{(4)}{=} q_m \bar{q}_m \sum_{k \ge 2} k(k - 1)\bar{q}_m^{k-2} + q_m \sum_{k \ge 1} k\bar{q}_m^{k-1} - \frac{1}{q_m^2} \\ &\stackrel{(5)}{=} q_m \bar{q}_m \left(\sum_{k \ge 0} \bar{q}_m^k\right)' + q_m \left(\sum_{k \ge 0} \bar{q}_m^k\right)' - \frac{1}{q_m^2} \\ &\stackrel{(6)}{=} q_m \bar{q}_m (-1) \frac{-2}{(1 - \bar{q}_m)^3} + \frac{1}{q_m} - \frac{1}{q_m^2} \\ &= \frac{2 - 2q_m}{q_m^2} - \frac{1 - q_m}{q_m^2} = \frac{1 - q_m}{q_m^2}. \end{aligned}$$

In (4), the index can be shifted to  $k \ge 2$  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

$$1 - p = (1 - q_m)^N$$

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

$$N = \frac{\log(1-p)}{\log(1-q_m)}.$$
(2.25)

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 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  $C \subseteq \mathbb{R}^2$  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

$$\mathbb{E}(K_2) = \frac{1}{\left(\frac{1}{3}\right)^2} = 9$$

and the variance and standard deviation by

$$\mathbb{V}(K_2) = \frac{1 - \frac{1}{9}}{\left(\frac{1}{9}\right)^2} = 72$$
 and  $\sqrt{\mathbb{V}(K_2)} = 6\sqrt{2} \approx 8.49.$ 

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





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 2i 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.



(a) Truck and autonomous forklift.



(b) ZED Camera of Stereolabs [54].

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 SO<sub>3</sub> 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

$$u^* = \begin{pmatrix} \cos(-\alpha) & 0 & -\sin(-\alpha) \\ 0 & 1 & 0 \\ \sin(-\alpha) & 0 & \cos(-\alpha) \end{pmatrix} \begin{pmatrix} 0 \\ 0 \\ 1 \end{pmatrix} = \begin{pmatrix} -\sin(-\alpha) \\ 0 \\ \cos(-\alpha) \end{pmatrix}.$$

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 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 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  $\{e_l, e_w\}$  and the longest visible edge orthogonal to  $e_1$  and  $u^*$  is also in  $\{e_l, e_w\}$ .

To find the edge  $e_1$ , we first compute possible edge points in 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 kd-tree of the points in C (see Section 2.4.1 for more about kd-trees) and estimate the normals of the points. To further accelerate the computation, the point cloud 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 C'. The roughly 200,000 points in C reduce to about 5,000 points in C'. Now we search for edge points in the point cloud C'.







(c) Outliers of the point cloud (b) removed.

(d) Point cloud cut to the interesting region.

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 C instead of C'.

For each point p in  $\mathcal{C}'$ , 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 kd-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  $N_p^{\text{rest}} := N_p \setminus N_p^{\text{up}}$  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  $N_p^{\text{up}}$  have (approximately) the same direction  $v_p^{\text{up}}$ .
- (ii) The normals of the points in  $N_p^{\text{rest}}$  have (approximately) the same direction  $v_p^{\text{rest}}$
- (iii) The vectors  $v_p^{\text{up}}$  and  $v_p^{\text{rest}}$  are (approximately) orthogonal.
- (iv) The sets  $N_p^{\text{up}}$  and  $N_p^{\text{rest}}$  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

$$v_p^{\text{up}} = \frac{1}{|N_p^{\text{up}}|} \sum_{q \in N_p^{\text{up}}} n_q \quad \text{and} \quad v_p^{\text{rest}} = \frac{1}{|N_p^{\text{rest}}|} \sum_{q \in N_p^{\text{rest}}} n_q$$

where  $n_q$  is the estimated normal vector of a point q. The normals  $n_q$  in  $N_p^{\text{up}}$  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  $v_p^{\text{up}}$  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  $v_p^{\text{up}}$  and  $v_p^{\text{rest}}$  are not the zero-vector. Thus, we can compute the inner product

$$\left\langle \frac{1}{\|v_p^{\rm up}\|} v_p^{\rm up}, \frac{1}{\|v_p^{\rm rest}\|} v_p^{\rm rest} \right\rangle. \tag{3.1}$$

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)]

$$\cos \alpha = \frac{\langle v_p^{\mathrm{up}}, v_p^{\mathrm{rest}} \rangle}{\|v_p^{\mathrm{up}}\| \|v_p^{\mathrm{rest}}\|} = \left\langle \frac{1}{\|v_p^{\mathrm{up}}\|} v_p^{\mathrm{up}}, \frac{1}{\|v_p^{\mathrm{rest}}\|} v_p^{\mathrm{rest}} \right\rangle.$$

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  $N_p^{\text{up}}$  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  $C_{edge} \subseteq C'$  of edge points. The edge  $e_1$  is defined as the line passing through the most points in  $C_{edge}$ , 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  $C_{edge}$  for the longest edge (approximately) orthogonal to  $e_1$ , using a RANSAC approach. In particular, with RANSAC the line with the most votes in  $C_{edge}$  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  $C_{edge}$  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  $\frac{1}{2}(S_1 + S_2)$  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  $C_{edge}$ : 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

$$v_x^{\text{left}} := v_l, \qquad v_z^{\text{left}} := v_x^{\text{left}} \times v_w, \qquad v_y^{\text{left}} := v_z^{\text{left}} \times v_x^{\text{left}}.$$

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

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

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  $c_p = \frac{1}{n} \sum_{p' \in N_p} p'$  of the neighbors of p and measure the distance  $d(p, c_p)$ . 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  $d(p, c_p) = ||c_p - p||_2$ . 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 kd-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

$$||o^{\text{est}} - o^{\text{true}}||_2 + ||v_y^{\text{est}} - v_y^{\text{true}}||_2 + ||v_x^{\text{est}} - v_x^{\text{true}}||_2$$

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.



(a) Truck viewed from far away. (b) Close up of 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  $T \in SE_3$  that relates the poses  $P_{p_1}$  and  $P_{p_2}$  of the parts  $p_1$  and  $p_2$  of the truck as  $T(P_{p_1}) = P_{p_2}$ . When viewing the wheels of the truck as right circular cylinders, the class of proper rigid transformations  $\mathcal{T}_{P_w}^{P_w'}$  for two wheels w and w' of Definition 2.1.17 contains more than one element of SE<sub>3</sub>. Here, we ignore this problem of symmetry that would result in transformations  $T, T' \in SE_3$  with  $T \neq T'$  but  $T \sim_{P_1} T'$  according to Definition 2.1.16, i.e. T and T' 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  $T(P_{p_1}) = P_{p_2}$  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  $T \in SE_3$  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  $(R, t) \in SE_3$ with  $t = (0, 0, 0)^T \in \mathbb{R}^3$  and  $R = I_3 \in SO_3$ .

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 SE<sub>3</sub>. 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 ycoordinate, 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  $s_{t'}$  for  $t' < t_0$  with the current time step  $t_0$  that are not deleted from the graph. For a part p with the old pose  $T_p^{\text{old}} = (R_p^{\text{old}}, t_p^{\text{old}}) \in \text{SE}_3$ , 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

$$t_p^{\text{new}} := w t_p^{\text{est}} + (1 - w) t_p^{\text{old}}$$

$$(3.2)$$

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

$$w(n) = \lambda_w^n (w_0 - c) + c, \tag{3.3}$$

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  $c \in [0, w_0)$  the lower bound of w. According to the computation

$$w(n) = \lambda_w^n (w_0 - c) + c$$
$$\lambda_w w(n) = \lambda_w^{n+1} (w_0 - c) + \lambda_w c$$
$$\lambda_w^{n+1} (w_0 - c) = \lambda_w w(n) - \lambda_w c$$

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

 $w(n+1) = \lambda_w w(n) + (1 - \lambda_w)c.$ 

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  $SO_3$ . 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  $Q = UDV^T$  with  $U, V \in O_3$  and the diagonal matrix  $D = \text{diag}(\alpha_1, \alpha_2, \alpha_3)$  with  $\alpha_1 \ge \alpha_2 \ge \alpha_3 \ge 0$ . The averaged rotation is then given by  $R_p^{\text{new}} = USV^T$  with  $S = \text{diag}(1, 1, \text{sgn}(\det(U) \det(V)))$ . The matrix  $R_p^{\text{new}}$  is indeed in SO<sub>3</sub>.

The new poses  $T_p^{\text{new}} = (R_p^{\text{new}}, t_p^{\text{new}})$  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 3.16c, 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 3.16f 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 SE<sub>3</sub> defined in Theorem 2.1.31. We define the *cumulated model error*  $\operatorname{err}^{(t)}$  in time step t as

$$\operatorname{err}^{(t)} := \sum_{p \in P} d(T_p^{(t)}, T_p^{\operatorname{true}}),$$

where P denotes the set of all parts of the truck,  $T_p^{(t)} \in SE_3$  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  $\operatorname{err}^{(t)}$  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  $\mathcal{N}(0, 0.5^2)$  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  $\mathcal{N}(0, 0.02^2)$ 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.



Cumulated Model Error Over 125 Time Steps for Different Values of construction noise (translation)

(b) Sensor estimation error.

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

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(a) Model error depending on the updating factor  $\lambda_c$  for the noise of the construction factors. Cumulated Model Error Over 125 Time Steps for Different Values of updating factor (weights)



(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 3.17b was created for the same situation but displays the error of the sensor estimation measured as  $d(T_{s_t}^{\text{est}}, T_{s_t}^{\text{true}})$  with the metric d on SE<sub>3</sub> 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 SE<sub>3</sub>. We have seen that SO<sub>3</sub> and SE<sub>3</sub> 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  $\mathbb{R}^{3\times3}$  an explicit representation as Rodrigues' formula. For manifold optimization 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
2
3
    import time
4
    def run_loading_edge_detection(filename, inputs):
         .....
5
\mathbf{6}
         Run the loading edge detection.
7
        Parameters
8
9
        filename : str
            Name of the file containing the point cloud.
10
         inputs : dict
11
            Dictionary containing some custom parameter values.
12
13
        Returns
14
15
        loading_edge : ndarray of shape (2,3)
            Two points that define the loading edge.
16
17
         other_edge : ndarray of shape (2,3)
             Two points that define the second edge.
18
        running_time : float
19
20
            Duration of the loading edge detection.
21
        num_edges_found : int
             0, 1, or 2, dependent on the number of edges found.
22
         .....
23
24
        # read in the point cloud
        pcd = o3d.io.read_point_cloud(filename)
25
         # define an instance of the class with the desired parameters
26
27
        led = LoadingEdgeDetection(o3d_cloud=pcd, **inputs)
         # perform voxel downsampling
28
         voxel_downsampled_pcd = led.voxel_downsample_pcd(led.pcd, voxel_size=0.015)
29
30
         # remove the outliers
31
         final_pcd = led.remove_outliers(voxel_downsampled_pcd, nb_points=30, radius=0.05)
32
         # find the loading edge
         loading_edge, other_edge, running_time, num_edges_found = led.find_two_edges(final_pcd)
33
        return loading_edge, other_edge, running_time, num_edges_found
34
    class LoadingEdgeDetection:
35
        def __init__(
36
37
             self,
             o3d_cloud,
38
             est_angle_camera=32.5,
39
40
             est_height_camera=2.05,
41
             est_height_loading_platform=1.35,
             search_width=0.5,
42
43
             tol_orthogonal=0.3,
```

44		max_lines=5,
45		uniform_downsample_factor=40,
46	):	
47		<i>nnn</i>
48		Constructor.
49		Parameters
4 <i>3</i>		
50		and aloud a grant doint floud
51		Dist ind when the local construction
52		Forni cioua, minere ine iouarity euge snouta de aelectea.
53		est_camera_angle : jtoat, oprional
54		Estimatea camera angle in aegree to the norizontal plane. The aefault is 32.5.
55		est_height_camera : float, optional
56		Estimated height of the camera above the ground in meters. The default is 2.05.
57		est_height_loading_platform : float, optional
58		Estimated height of the loading platform above the ground in meters. The default is 1.35.
59		search_width : float, optional
60		Width of the remaining point cloud, when cutting the pcd above and below
61		the estimated plane of the loading platform. The default is 0.5.
62		tol_orthogonal : float, optional
63		Tolerance when searching for orthogonal lines. Has to be between 0 and 1.
64		The default is 0.3.
65		max_lines : int, optional
66		Maximum number of lines to be found in 'line_ransac'. The default is 5.
67		uniform_downsample_factor : int, optional
68		Downsample factor when reducing the number of points for the edge point search.
69		The default is 40.
70		Attributes
71		
72		ncd : open3d PointCloud
73		Original point cloud of the scene.
74		noints · ndarray of shane (num noints num dimensions)
75		Points of the original point cloud as nummy array
76		est beight camera - float
77		Estimated height of the camera above the argund in meters. Used to estimate the mose
78		of the loading mlatform
70		est beight loading platform · float
80		Estimated beight of the loading mlatform above the around in metame. Used to estimate
01		the mass of the loading mattern
01		ine pose of the touting plutform.
82		up_vector : maintait of shape (3,)
83		Estimated up-bector of the scenery. Computed with the estimated angle of the camera.
84		point_tourng_pittjorm_oetou_camera : naurray of shape (3,)
85		Forth in the estimated height of the todard platform, vertically below the camera.
86		computed by the estimated up-vector and neight of the camera and the loading platform.
87		search_with : float
88		width of the remaining point cloud, when cutting the pcd above and below the estimated
89		plane of the loading platform. Smaller search width speeds up the computation and
90		eliminates other edge points. Can be decreased, if angle and height estimates are good.
91		tol_orthogonal : float in [0, 1]
92		Tolerance when searching for orthogonal lines. It holds:
93		$v$ is orthogonal to $w \iff v \mathcal{C} w = 0$ .
94		Normalized vectors $v$ and $w$ are approximately orthogonal, if $v \mathrel{@} w < tol_orthogonal.$
95		max_lines : int
96		Maximum number of lines to be found in 'line_ransac' before the algorithm stops.
97		Larger max_lines is more likely to find a second edge,
98		but also more likely to declare a line as an edge, that is not an edge.
99		origin : ndarray of shape (3,)
100		Origin of the world coordinate frame, defined in the method find_origin(). The origin of
101		the world frame is defined as the beginning of the left loading edge. If the truck is seen
102		from the right side, 'origin' states the beginning of the right loading edge.
103		coordinate_frame : dict
104		Contains the directions of the coordinate axes of the world frame, defined in the method

```
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```

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```
rotate_coordinate_frame(). The keys are 'x', '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 plane: 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):
```

## A. Code Loading Edge Detection

158	Computes the indices of the points inside or outside the cut region for cut_pcd_with_plane().
159	Parameters
160	
161	pcd_np : ndarray of shape (num_points, num_dimensions)
162	Point cloud as numpy array.
163	weights : list of float
164	Weights $a,b,c,d$ of the plane: $a*x + b*y + c*z + d = 0$ .
165	dist_to_original_plane : float
166	Distance of the planar cut to the original plane.
167	inside : bool
168	If True, the points inside the cut region are returned.
169	If False, the points outside the cut region are returned.
170	Returns
171	
172	list of int
173	Indices of the points inside or outside the cut region.
174	"""
175	# transpose the array to ease computation (pcd_np[0] is a vector of all first coordinates)
176	<pre>pcd_np = pcd_np.T</pre>
177	# compute the values of the plane equation for all points as a numpu array
178	<pre>values = weights[0] * pcd_np[0] + weights[1] * pcd_np[1] + weights[2] * pcd_np[2] + weights[3]</pre>
179	# cut in both directions
180	above = values $>$ np.abs(dist to original plane)
181	below = values < -np.abs(dist_to_original_plane)
199	t return the indices of the naints inside or outside the cut region
102	if not ingide.
104	not inside.
184	electin [1 for 1 in range(ren(above)) if above[1] of berow[1]]
180	erse:
180	recurn [1 101 1 in range(ren(above)) if not above[1] and not perow[1]]
187	""
188	Additional functions, not shown here:
189	def divide_inlier_outlier(self, pcd, inlier_ind, color)
190	def voxel_downsample_pcd(self, pcd, voxel_size)
191	def remove_outliers(self, pcd, nb_points, radius)
192	def pcd_to_np(self, pcd)
193	def draw_point_cloud(self, list_of_pcds, show_normals, estimate_normals)
194	
195	**********************
196	# Truck Specific Functions
197	***************************************
198	<pre>def get_up_vector(self, camera_angle):</pre>
199	<i>ини</i>
200	Compute the estimated up-vector (up-direction of the scenery in the world frame) from the
201	estimated camera angle as np.array([np.sin(-camera_angle), 0, np.cos(-camera_angle)]).
202	Parameters
203	
204	camera_angle : float
205	Estimated camera angle in degrees to the horizontal plane.
206	Returns
207	
208	up_vector : ndarray of shape (3,)
209	Approximate up-vector.
210	nun -

```
# z-axis goes up (x to the front, y to the side)
211
212
              up_vector_camera = np.array([0, 0, 1])
              # angles to rotate around (rotate around the y-axis) in radians
213
              angles = np.deg2rad(np.array([0, -camera_angle, 0]))
214
215
              # rotation matrix
216
              rot_matrix = o3d.geometry.get_rotation_matrix_from_axis_angle(angles)
217
              return rot_matrix @ up_vector_camera
          def get_point_loading_platform_below_camera(self):
218
219
              Computes the point below (in the direction of the estimated up-vector) the camera
220
221
              that lies in the estimated height of the loading platform.
              Returns
222
223
224
              point : ndarray of shape (3,)
                  Estimated position of the point in the height of the loading platform, below the camera.
225
                  Given in sensor frame coordinates.
226
              .....
227
              # the camera defines the origin of the sensor frame
228
229
              pos_camera = np.array([0, 0, 0])
230
              # height difference between camera and loading platform
231
              height_diff_camera_loading_platform = (
                  self.est_height_camera - self.est_height_loading_platform
232
233
              )
234
              return pos_camera - height_diff_camera_loading_platform * self.up_vector
235
          def cut_pcd_around_loading_platform(self, pcd):
236
237
              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.
238
239
              Parameters
240
              _____
              pcd : open3d PointCloud
241
242
                 Input point cloud.
243
              Returns
244
245
              open3d PointCloud
                 Output point cloud.
246
              .....
247
              # equation of the plane: nX = nP \rightarrow nX - nP = 0
248
249
              # the up-vector defines the normal vector n of the plane
              weights = list(self.up_vector)
250
251
              # append -nP to the weights
252
              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
253
              return self.cut_pcd_with_plane(pcd, weights, self.search_width / 2, True)
254
          def find_origin(self, P_line1, P_line2):
255
256
257
              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
258
259
              defined by the other line and the vertical direction.
```

## A. Code Loading Edge Detection

260	Then we take the average of the two intersection points.
261	Parameters
262	
263	P line1 : ndarray of shape (2.3)
260	The main set of define the first line
204	Dimo pointo internati de jone de jone dente.
265	r_triez: nuarray of shape (2,3)
266	Iwo points that define the second line.
267	set_origin : bool, optional
268	If Irue, the origin of the coordinate frame is set as an attribute of the class.
269	The default is False.
270	Returns
271	
272	origin : ndarray of shape (3,)
273	Found origin of the coordinate frame.
274	"""
275	# get the intersection points
276	<pre>intersection_point1 = self.intersect_line_plane(</pre>
277	P_line1, np.array([P_line2[0], P_line2[1], P_line2[0] + self.up_vector])
278	
279	intersection_point2 = self.intersect_line_plane(
280	P_line2, np.array([P_line1[0], P_line1[1], P_line1[0] + self.up_vector])
281	)
282	# commute the origin and set it as a class attribute
282	Triging (intersection point) + intersection point2) / 2
200	alf origin - (intersection_point) + intersection_point2) / 2
284	sen origin - origin
285	return origin
286 def	<pre>find_direction(self, edge_pcd, origin, P):</pre>
287	""
288	This function is used to find the true direction of an axis (deciding between vec and -vec)
289	to find the correct coordinate frame. The correct direction is the vector from the origin
290	to the endpoint of P that is further away from the origin. This is done by finding the endpoints of the line and commaring the distances of these endpoints to the origin
201	
292	Parameters
293	
294	eage_pca : opensa Pornicioua
295	Point cloud that contains the points on the eage.
296	Inis point cloud should already be the largest cluster of a found eage.
297	origin : ndarray of shape (3,)
298	Urigin of the coordinate frame.
299	P: ndarray of shape (2,3)
300	Two points that define the line.
301	The line defined by P does not have to be parallel to the line defined by the endpoints.
302	The two endpoints might just give a rough approximation of the direction of the line.
303	Returns
304	
305	P: ndarray of shape (2,3)
306	Two points that define the line.
307	They are the same points as the input P, but the order of the points might be switched.
308	
309	# direction vector of the line
310	vec = P[1] - P[0]
311	# get the approximate endpoints of the line
312	endpoint1, endpoint2 = self.find_endpoints(edge_pcd, max_iter=100)
313	# compare the distances of the endpoints to the origin

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 $366 \\ 367$ 

```
dist_origin_endpoint1 = np.linalg.norm(endpoint1 - origin)
   dist_origin_endpoint2 = np.linalg.norm(endpoint2 - origin)
    if dist_origin_endpoint1 > dist_origin_endpoint2:
        \ensuremath{\texttt{\#}} 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
```

```
369
             # set the axes and the rotation matrix as attributes of the class
370
             self.coordinate_frame["x"] = new_x_axis
             self.coordinate_frame["y"] = new_y_axis
371
             self.coordinate_frame["z"] = new_z_axis
372
             self.rot_matrix_to_coordinate_frame = rot_matrix
373
374
             return rot_matrix
375
         376
         # Edge Detection
         377
378
         def find_two_edges(self, pcd):
379
380
             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
381
382
             are found, the function 'differentiate_edges' is used to distinguish the loading edge from
383
             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
384
             set such that the loading platform is located in the first quadrant of the xy-plane.
385
             The two edges are then drawn together with the rest of the point cloud.
386
387
             Parameters
388
389
             pcd : open3d PointCloud
390
                Point cloud, where the edges should be detected.
391
             Returns
392
393
             edge1_P : ndarray of shape (2,3)
394
                Two points that define the loading edge.
             edge2_P : ndarray of shape (2,3)
395
396
                Two points that define the second edge.
397
             time : float
                Duration of the loading edge detection.
398
             num_edges_found : int
399
400
                0, 1, or 2, dependent on the number of edges found.
             .....
401
             # start the timer
402
             start = time.time()
403
             # find the longest edge in the point cloud
404
405
             edge1_P, edge1_pcd, edge1_ind, edge_points_pcd, found_longest_edge = self.find_longest_edge(
406
                pcd
             )
407
408
             # if there are enough edge points to find an edge, we look for a second edge
             if found_longest_edge:
409
410
                 # find a second edge that is orthogonal to the first edge
                 edge2_P, second_line_found = self.find_orthogonal_line(
411
                    edge_points_pcd, P_original=edge1_P, max_lines=self.max_lines
412
                )
413
                 # if a second edge is found, the coordinate frame can be set
414
415
                 if second_line_found:
                    num_edges_found = 2
416
417
                    origin = self.find_origin(edge1_P, edge2_P)
                    edge2_ind = self.get_points_on_line(pcd, edge2_P)
418
                    edge2_pcd, _ = self.divide_inlier_outlier(pcd, edge2_ind, color=[0, 1, 0])
419
                    edge1_P = self.find_direction(edge1_pcd, origin, edge1_P)
420
```

```
edge2_P = self.find_direction(edge2_pcd, origin, edge2_P)
421
422
                      # define the direction of the up-vector (currently it is up or down)
423
                      # 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°
424
                      # if the angle is larger than 90°, the up-vector is pointing downwards
425
                      if self.up_vector @ (np.array([0, 0, 0]) - origin) < 0:
426
427
                          self.up_vector *= -1
428
                      # check if edge1 or edge2 is the loading edge
                      _, _, correct_order = self.differentiate_edges(pcd, edge1_P, edge2_P)
429
                      # switch edge1 and edge2 if necessary
430
                      if not correct_order:
431
                          edge1_P, edge2_P = edge2_P, edge1_P
432
433
                          edge1_pcd, edge2_pcd = edge2_pcd, edge1_pcd
                          edge1_ind, edge2_ind = edge2_ind, edge1_ind
434
435
                      # get the rotation matrix to rotate the coordinate frame
                      rot_matrix, axis_loading_edge = self.rotate_coordinate_frame(
436
                          edge1_P[1] - edge1_P[0], edge2_P[1] - edge2_P[0]
437
438
                      end = time.time()
439
440
                      # here would be the place to visualize the results
                      return edge1_P, edge2_P, end - start, num_edges_found
441
                  else:
442
443
                      num_edges_found = 1
444
                      # if only one edge is found,
                      # the second horizontal direction is determined by the first edge and the up-vector
445
                      edge1_P = self.find_endpoints(edge1_pcd)
446
                      # the origin is one endpoint of the first edge
447
                      origin = edge1_P[0]
448
                      \ensuremath{\textit{\#}} get the direction of the second axis
449
                      potential_second_axis = np.cross(edge1_P[1] - edge1_P[0], self.up_vector)
450
                      # the loading platform locates on the side of the first edge, where more points are
451
                      lifting_factor = 0.3
452
453
                      lift1 = lifting_factor * potential_second_axis
454
                      edge1_P_lifted1 = edge1_P + lift1
                      lift2 = lifting_factor * (-potential_second_axis)
455
                      edge1_P_lifted2 = edge1_P + lift2
456
                      num_votes_lifted_1 = len(
457
458
                          self.get_points_on_line(
                              pcd, edge1_P_lifted1, tol=0.1, cluster=False, between_points=True
459
                          )
460
                      )
461
462
                      num_votes_lifted_2 = len(
463
                          self.get_points_on_line(
464
                              pcd, edge1_P_lifted2, tol=0.1, cluster=False, between_points=True
                          )
465
                      )
466
                      if num_votes_lifted_1 < num_votes_lifted_2:</pre>
467
                          second_axis = -potential_second_axis
468
                      else:
469
                          second_axis = potential_second_axis
470
```

471	<pre>second_axis /= np.linalg.norm(second_axis)</pre>
472	# act the rotation matrix to rotate the coordinate frame
473	rot_matrix = self.rotate_coordinate_frame(edge1_P[1] - edge1_P[0], second_axis)
474	<pre>end = time.time()</pre>
475	# here would be the place to visualize the results
476	<pre>edge2_P = np.array([origin, origin + second_axis])</pre>
477	return (edge1_P, edge2_P, end - start, num_edges_found)
478 de	<pre>f find_longest_edge(self, pcd):     """</pre>
479	 Tigd dby langest adapting a stick along. First we find dby adapting and dby a line dby
480 481	contains many of these edge points.
482	Parameters
483	ncd · omen3d Point(1)and
485	Point cloud, where the edge should be detected.
486	Returns
487	
488	eage_P : naarray of shape (2,3)
489	The endpoints of the dominant edge in the point cloud.
490	eage_pca : opensa rornicioua
491	Point cloud that contains the points on the dominant edge.
492	cuge ind , is to j inc. Indices of the moints on the dominant edge
493	indices of the points on the administrative edge.
494	Point cloud that contains the edge points
496	edue found : bool
497	Indicates if the algorithm found an edge.
498	nun
499	edge_found = True
500	# cut the point cloud above and below the estimated plane of the loading platform
501	<pre>pcd_cut = self.cut_pcd_around_loading_platform(pcd)</pre>
502	# find the edge points with the normals of the points
503	<pre>edge_points_ind = self.find_edge_points_normals(pcd_cut)</pre>
504	# check if enough edge points were found
505	if len(edge_points_ind) < 3:
506	<pre>edge_found = False</pre>
507	return None, None, None, edge_found
508	# divide the cut point cloud into edge points and non-edge points
509	<pre>edge_points_pcd, _ = self.divide_inlier_outlier(pcd_cut, edge_points_ind)</pre>
510	# find a line that contains many of these edge points
511	<pre>P, ransac_completed = self.line_ransac(edge_points_pcd)</pre>
512	if ransac_completed:
513	# find the points of the original point cloud on this line
514	<pre>edge_ind = self.get_points_on_line(pcd, P, between_points=False)</pre>
515	# arrive the original point cloud into edge points and non-edge points adroped = colf divide inlice outlier(red edge ind color=[1, 0, 0])
910	eage_pcu, _ = sell.alviae_initer_outlier(pca, eage_ina, color=[1, 0, 0])

```
edge_P = self.find_endpoints(edge_pcd)
517
518
                  return edge_P, edge_pcd, edge_ind, edge_points_pcd, edge_found
519
              else:
                  edge_found = False
520
521
                  return None, None, None, None, edge_found
          def find_edge_points_normals(self, pcd, radius=0.1):
522
523
             Find edge points of a point cloud by computing the normals of the neighbors of each point.
524
525
             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
526
             orthogonal to the vertical direction, the point is considered an edge point.
527
528
             Uses the function 'cluster_normals' to cluster the normals.
             Parameters
529
530
531
             pcd : open3d PointCloud
                 Point cloud, where the edge points should be found.
532
533
              radius : float, optional
                 Radius around a point, where the kd-tree looks for neighbors. The default is 0.1.
534
535
             Returns
536
537
              edge_points : list of int
538
                  Indices of the edge points.
              .....
539
540
              # compute the kd-tree of the point cloud to enable fast neighbor search
             kdtree = o3d.geometry.KDTreeFlann(pcd)
541
              # compute the normals of the point cloud
542
             pcd.estimate_normals()
543
544
              # downsample the point cloud to speed up computation
             downsampled_pcd = pcd.uniform_down_sample(self.uniform_downsample_factor)
545
              edge_points_ind = []
546
             for i in range(len(downsampled_pcd.points)):
547
                  # find the neighbors of the point
548
                  _, point_indices, _ = kdtree.search_radius_vector_3d(downsampled_pcd.points[i], radius)
549
                  # get the point cloud of the neighbors
550
551
                  neighbor_pcd = pcd.select_by_index(point_indices)
                  # cluster the normals of the neighbors into two groups
552
553
                  orthogonal, _ = self.cluster_normals(neighbor_pcd)
                  # if the normals could be clustered into two groups that are approximately orthogonal,
554
555
                  # the point is considered an edge point
556
                  if orthogonal:
                      edge_points_ind.append(self.uniform_downsample_factor * i)
557
558
             return edge_points_ind
          def cluster_normals(self, pcd, tol_cluster=0.4):
559
560
             Helper function for 'find_edge_points_normals'. Clusters the normals of a point cloud into
561
              two groups. If the mean of one group is nearly parallel to the vertical direction, and the
562
563
             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.
564
565
             Parameters
```

566	
567	ncd · onen3d Point(1) and
568	Point cloud, where the normals should be clustered. Usually the neighbors of a point.
569	tol_cluster : float, optional
570	Tolerance of the inner product of the mean of the two clusters and the vertical direction.
571	If the important $i \in O$ , the past is option and to the un-vector. The default is $O/$
571	If the time product is 0, the rest is orthogonal to the up-vector. The dejuant is 0.4.
572	Keturns
573	
574	orthogonal : bool
575	The two clusters are approximately orthogonal to each other.
576	contain most : ndownay of change (2)
570	center_rest . numruy of snape (3,)
577	Mean of the cluster that is that is not the vertical direction.
578	If the two clusters are not approximately orthogonal, None is returned.
579	""
580	# get the normals of the points in the point cloud
581	normals = nn asarray(ncd normals)
561	normais – np.asarray(ped.normais)
E 9.9	# telemance for the length of the difference of the respect on the section
582	# loterance for the length of the alfference of the normals and the vertical alfection
583	tol_parallel = 0.3
584	# all normals that are NOT approximately vertical (parallel to up_vector)
585	rest = normals[
586	nn logical and(
087	np.iinaig.noim(noimais - seii.up_vector, axis-i) / toi_paraiter,
588	np.linalg.norm(normals + self.up_vector, axis=1) > tol_parallel,
589	)
590	]
591	# if there are too little normals of one kind, this is not an edge point
592	if $4 + len(rest) > len(normals) and 4 + len(rest) < 3 + len(normals):$
502	# compute the mean and the mean of the mean
093	# compute the mean and the norm of the rest
594	center_rest = np.sum(rest, axis=0) / len(rest)
595	<pre>norm_rest = np.linalg.norm(center_rest)</pre>
596	# tolerance for the length of the mean of the rest
597	tol_norm = 0.4
	-
598	# if the norm is too short, the normals are too far away from each other $\rightarrow$ return False
-00	if your work to be not by the normalize are bob jur away from each bonch - repair rates
599	11 norm_rest > to1_norm:
600	# normalize the mean of the rest
601	center_rest /= norm_rest
602	# compute the inner product of the mean of the rest and the vertical direction
603	inner_product = center_rest @ self.up_vector
604	# if the inner product of the centers is close to zero.
001	<pre># the electronic processing one service or the constant</pre>
605	# the clusterea normals are nearly orthogonal
606	if np.abs(inner_product) < tol_cluster:
607	return True, center_rest
608	return False, center_rest
609	return False, None
610	def differentiate_edges(self, pcd, edge1_P, edge2_P, lifting_vector=None):
611	
619	Differentiate between a loading edge and the edge between loading mlatform and mean wall
014	buy difference between a bounding cage and one cage between bounding platform and teat with
613	ney aijjerence: above the loaarng eage, there should be no points -> translate the edge a bit
614	in the direction of the up_vector and check, which line hits less points -> loading edge
615	Parameters
616	
-	

```
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.
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:</pre>
    # 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:</pre>
    # edge1 is the loading edge
    loading_edge = edge1_P
```

 $\begin{array}{c} 617 \\ 618 \end{array}$ 

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671

673	other_edge = edge2_P
674	correct_order = True
675	else:
676	# edge2 is the loading edge
677	loading_edge = edge2_P
678	other_edge = edge1_P
679	correct_order = False
680	return loading_edge, other_edge, correct_order
681	*****
682	# Geometric Functions
683	***************************************
684	<pre>def line_ransac(self, pcd, tol=0.08, max_iter=100):</pre>
685	"""
686	Find a line in a point cloud by RANSAC. Typical usecase: The given point cloud consists of
687	previously detected edge points, and we want to find the dominant edge, i.e. the edge
688	containing the most points.
689	Parameters
690	
691	pcd : open3d PointCloud
692	Point cloud, where the line should be detected.
693	Usually this point cloud consists of previously detected edge points.
694	tol : float, optional
695	Tolerance of the distance between a point and the line. The default is 0.08.
696	max_iter : int, optional
697	Maximum number of iterations of the RANSAC algorithm. The default is 100.
698	Returns
699	
700	best_line : ndarray of shape (2,3)
701	Two points that define the line.
702	finished_ransac : bool
703	States if there are enough points in the point cloud to perform RANSAC.
704	
705	if len(pcd.points) > 1:
706	# initialize the best line and the number of votes
707	<pre>best_votes = 0</pre>
708	<pre>best_line = None</pre>
709	for in range(max iter):
710	# sample the points and get the votes for this line
711	P1, P2, votes = self.line_ransac_votes(pcd, tol)
(12	# upuale the vest line and the number of votes if necessary
713	li votes > best_votes:
714 715	best line = np.arrav([P1, P2])
716	return best_line, True
717	else:
718	return None, False
719	<pre>def line_ransac_votes(self, pcd, tol, P=None):</pre>
720	
721	Helper function for 'line_ransac'. Counts the number of points that lie within some small
722	tolerance around a line. If P is None, two random points are sampled from pcd.
792	Damamaters
123	r ut unever s
```
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:
       P1 = P[0]
        P2 = 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</pre>
   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)</pre>
   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= {\it True, only the points between the two given points are returned}. \\
   Parameters
     -----
```

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## A. Code Loading Edge Detection

775 776	pcd : open3d PointCloud Point cloud, where the points on the line should be found.
777	P: ndarray of shape (2,3)
778	Two points that define the line.
779	tol : float, optional
780	Tolerance of the distance between a point and the line. The default is 0.05.
781	cluster : bool, optional
782	If True, the indices of the largest cluster of points on the line are returned.
783	If False, all the indices of the points on the line are returned.
784	Only relevant, if between_points=False. The default is True.
785	between_points : bool, optional
786	If True, only the points between the two given points are returned.
787	If False, all the points on the line are returned. The default is False.
788	Returns
789	
790	points_on_line_ind : list of int
791	Indices of the points on the line.
792	
793	# get the two points that define the line
794	PI, PZ = P[0], P[1]
705	# net the distance of all moints to the first moint of the line
796	"got the noises = ord noises - $P1$
150	
797	# get the vector of the line and its length
798	vec_line = P2 - P1
799	<pre>original_len_line = np.linalg.norm(vec_line)</pre>
800	# normalize the vector of the line
801	vec_line /= original_len_line
802	# meanact the mectane to the manute anto the mectan of the line
002	* project the dectors to the points onto the dector of the time
803	<pre>projected_length = vecs_to_points @ vec_line</pre>
802 803	<pre>projected_length = vecs_to_points @ vec_line if between points:</pre>
803 804 805	<pre>if between_points: # consider only points between P1 and P2</pre>
802 803 804 805 806	<pre># project the becors to the points onto the bector of the time 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 Puthagorean theorem</pre>
802 803 804 805 806 807	<pre>w project the vectors to the points onto the vector of the the 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(</pre>
802 803 804 805 806 807 808	<pre>w project the vectors to the points onto the vector of the the 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</pre>
802 803 804 805 806 807 808 809	<pre>w project the vectors to the points onto the vector of the time 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     )</pre>
802 803 804 805 806 807 808 809	<pre>w project the vectors to the points onto the vector of the the 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     )</pre>
802 803 804 805 806 807 808 809 810	<pre># project the bectors to the points onto the bector of the the 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</pre>
802 803 804 805 806 807 808 809 810 811	<pre># project the bectors to the points onto the bector of the the 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 &lt; 0] = 0</pre>
802 803 804 805 806 807 808 809 810 811 812	<pre># project the vectors to the points onto the vector of the the 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 &lt; 0] = 0 dist_to_line = np.sqrt(squared_dist)</pre>
802 803 804 805 806 807 808 809 810 811 812 813	<pre># project the vectors to the points onto the vector of the the 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)     points_on_line_ind = []</pre>
802 803 804 805 806 807 808 809 810 811 812 813 814	<pre># project the vectors to the points onto the vector of the the 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 &lt; 0] = 0 dist_to_line = np.sqrt(squared_dist) points_on_line_ind = [] for i in range(len(projected_length)):</pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815	<pre># projected_length = vecs_to_points onto the vector of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)     points_on_line_ind = []     for i in range(len(projected_length)):         if (</pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816	<pre># projected_length = vecs_to_points onto the vector of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0 </pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817	<pre># projected_length = vecs_to_points onto the vector of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line </pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817 818	<pre># projected_length = vecs_to_points onto the vector of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol </pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817 818 819	<pre># projected_length = vecs_to_points onto the vector of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol         ): </pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817 818 819 820	<pre># projected_length = vecs_to_points onto the vector of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol         ):             points_on_line_ind.append(i) </pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817 818 819 820 821	<pre># project the bettors to the points onto the bettor of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol         ):             points_on_line_ind.append(i)     return points_on_line_ind</pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817 818 819 820 821	<pre># project the bettors to the points onto the bettor of the time 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol         ):             points_on_line_ind.append(i)     return points_on_line_ind</pre>
802 803 804 805 806 807 808 809 810 811 812 813 814 815 816 817 818 819 820 821 822 802	<pre># project the bettors to the points onto the bettor of the time projected_length = vecs_to_points 0 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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[ind.append(i)         return points_on_line_ind.eppend(i)  else:     # consider all noietto </pre>
802         803         804         805         806         807         808         809         810         811         812         813         814         815         816         817         818         819         820         821         822         823	<pre># project the dectors to the points onto the bettor of the time project d_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 &lt; 0] = 0 dist_to_line = np.sqrt(squared_dist)  points_on_line_ind = [] for i in range(len(projected_length)):     if (         projected_length[i] &gt; 0         and projected_length[i] &lt; original_len_line         and dist_to_line[i] &lt; tol         ):             points_on_line_ind else:         # consider all points         # consider allenee         # conset points</pre>
802         803         804         805         806         807         808         809         810         811         812         813         814         815         816         817         818         819         820         821         822         823         824	<pre># project the dectors to the points onto the bettor of the time project d_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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol         ):             points_on_line_ind else:         # consider all points         # consider all points         # compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points to the line with the Pythagorean theorem         compute the distance of the points     } } </pre>
802         803         804         805         806         807         808         809         810         811         812         813         814         815         816         817         818         819         820         821         822         823         824         825	<pre># project the vectors to the points onto the vector of the time project d_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 &lt; 0] = 0 dist_to_line = np.sqrt(squared_dist)  points_on_line_ind = [] for i in range(len(projected_length)):     if (         projected_length[i] &gt; 0         and projected_length[i] &lt; original_len_line         and dist_to_line[id &lt; tol     ):         points_on_line_ind else:     # consider all points     # 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[i] </pre>
802         803         804         805         806         807         808         809         810         811         812         813         814         815         816         817         818         819         820         821         822         823         824         825         826         827	<pre># project the vectors to the points onto the vector of the time project d_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 &lt; 0] = 0     dist_to_line = np.sqrt(squared_dist)      points_on_line_ind = []     for i in range(len(projected_length)):         if (             projected_length[i] &gt; 0             and projected_length[i] &lt; original_len_line             and dist_to_line[i] &lt; tol         ):             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     ) </pre>

```
828
                  # avoid numerical errors and get the distance to the line
                  squared_dist[squared_dist < 0] = 0</pre>
829
                  dist_to_line = np.sqrt(squared_dist)
830
                  # get the indices of the points that lie within the tolerance
831
                  points_on_line_ind = np.where(dist_to_line < tol)[0]</pre>
832
                  # return the indices of the largest cluster
833
834
                  if cluster:
                      return self.find_largest_cluster(pcd, points_on_line_ind)
835
836
                  else:
837
                      return points_on_line_ind
838
          def find_largest_cluster(self, pcd, pcd_ind=None, eps=0.5, min_points=10):
839
              Find the largest cluster of a point cloud.
840
841
              Parameters
842
              pcd : open3d PointCloud
843
                  Point cloud, where the largest cluster should be found.
844
              pcd_ind : list of int or None, optional
845
                  Indices of the points in pcd that we want to find a cluster in.
846
                  If None, we search for the largest cluster in the whole point cloud, e.g. if we want to
847
848
                  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.
849
850
              eps : float, optional
851
                 Epsilon for the DBSCAN algorithm. The default is 0.5.
              min_points : int, optional
852
853
                  Minimum number of points for a cluster. The default is 10.
              Returns
854
855
856
              inlier_ind : list of int
                 Indices of the points of the largest cluster.
857
858
              if pcd_ind is None:
859
                  # search for the largest cluster in the whole point cloud
860
                  pcd_ind = list(range(len(pcd.points)))
861
              else:
862
863
                  # or just in the points with the given indices
864
                  pcd = pcd.select_by_index(pcd_ind)
865
              # cluster the pcd using dbscan
              cluster_labels = np.array(pcd.cluster_dbscan(eps=eps, min_points=min_points))
866
              # points labeled -1: noise
867
              # get a list of all cluster labels
868
869
              unique_labels = np.unique(cluster_labels)
              # count for each cluster label the amount of points in that label
870
              num_points_per_cluster = [len(np.where(cluster_labels == i)[0]) for i in unique_labels]
871
              # If the two largest clusters have the same amount of points, we have to decide somehow
872
              # between them. Generally, this is bad, because then there is no unique 'largest' cluster
873
              # just take the first cluster \rightarrow [0] at the end
874
              # (the first [0] gives us the first entry of the tuple (array,) that we get from np.where)
875
              max_points_per_cluster = np.max(num_points_per_cluster)
876
              max_label = np.where(num_points_per_cluster == max_points_per_cluster)[0][0]
877
878
              # get the label of the largest cluster
              largest_cluster_label = int(unique_labels[max_label])
879
```

```
880
              # get all the indices of pcd_ind that correspond to the points of the largest cluster
881
              # in the larger point cloud that pcd is embedded in
882
              inlier_ind = [pcd_ind[i] for i in list(np.where(cluster_labels == largest_cluster_label)[0])]
              return inlier_ind
883
884
          def find_orthogonal_line(self, edge_pcd, P_original, tol=0.08, max_iter=100, max_lines=5):
885
886
              For a given line, find a line that is approximately orthogonal to it within a point cloud.
887
              The line is found by a version of RANSAC. At most max_lines lines are found.
888
              The first line that is orthogonal is returned.
              If no orthogonal line is found, the original line is returned, together with False.
889
890
              Parameters
891
              edge_pcd : open3d PointCloud
892
893
                  Point cloud, where the line should be detected.
894
                  Usually this point cloud consists of previously detected edge points.
              P_orthogonal : ndarray of shape (2,3)
895
                  Two points that define the line that is orthogonal to the line we are looking for.
896
              tol : float, optional
897
                  Tolerance of the distance between a point and the line. The default is 0.08.
898
899
              max_iter : int, optional
                  Maximum number of iterations of the RANSAC algorithm for one line. The default is 100.
900
901
              max_lines : int, optional
                  Maximum number of lines that are tried to find. The default is 5.
902
903
              Returns
904
905
              best_line : ndarray of shape (2,3)
                  Two points that define the line.
906
907
              orthogonal : bool
908
                  True, if an orthogonal line was found.
909
              # initialize the best line and the number of votes
910
              best_votes = 0
911
              best_line = P_original
912
              # define the original line
913
              line_original = P_original[1] - P_original[0]
914
              line_original /= np.linalg.norm(line_original)
915
916
              for j in range(max_lines):
917
                  for _ in range(max_iter):
                      # if the point cloud is too small, we cannot find a line
918
                      if len(edge_pcd.points) >= 2:
919
920
                          P1, P2, votes = self.line_ransac_votes(edge_pcd, tol)
921
                      else:
                          return P_original, False
922
923
                      # update the best line and the number of votes if necessary
924
                      if votes > best_votes:
925
                          best_votes = votes
                          best_line = np.array([P1, P2])
926
                  # compute the vector of the best line
927
                  vec_line = best_line[1] - best_line[0]
928
                  vec_line /= np.linalg.norm(vec_line)
929
930
                  # if the line is orthogonal to the other line and to the vertical direction,
931
                  # we have found the correct line
```

```
if (
932
933
                      np.abs(vec_line @ line_original) < self.tol_orthogonal</pre>
934
                      and np.abs(vec_line @ self.up_vector) < self.tol_orthogonal
935
                  ):
                      return best_line, True
936
937
                  else:
                      # delete the points on this line from the edge point cloud and start again
938
                      new_ind = self.get_points_on_line(edge_pcd, best_line, tol=0.05)
939
940
                      edge_pcd = edge_pcd.select_by_index(new_ind, invert=True)
941
                      best_votes = 0
                      best_line = P_original
942
943
              # if no orthogonal line is found within the maximum number of iterations,
944
              # return the original line and False
945
              return P_original, False
946
          def find_endpoints(self, edge_pcd, max_iter=100):
947
              Find the endpoints of a line, i.e. the points on the line that are the furthest away from each
948
              other. This is done by a variant of RANSAC. Usually a small number of iterations is enough,
949
              since this does not have to be the best pair of points, a close approximation suffices.
950
              If necessary, increase the number of iterations given by max_iter.
951
              Parameters
952
953
              edge_pcd : open3d PointCloud
954
                  Point cloud that contains the points on the edge.
955
956
                  This point cloud should already be the largest cluster of a found edge.
              max_iter : int, optional
957
958
                  Maximum number of iterations. The default is 100.
959
              Returns
960
961
              best_P1 : ndarray of shape (3,)
                 First endpoint.
962
              best_P2 : ndarray of shape (3,)
963
964
                 Second endpoint.
              .....
965
              # initialize best distance and best points
966
              best_dist = 0
967
968
              best_P1 = None
              best_P2 = None
969
970
              # find the best pair of points within max_iter iterations
971
              for i in range(max_iter):
                  ind1, ind2 = np.random.randint(len(edge_pcd.points), size=2)
972
973
                  P1 = edge_pcd.points[ind1]
                  P2 = edge_pcd.points[ind2]
974
                  new_dist = np.linalg.norm(P2 - P1)
975
                  if new dist > best dist:
976
977
                      best_dist = new_dist
                      best_P1 = P1
978
                      best_P2 = P2
979
              return best_P1, best_P2
980
          .....
981
          Additional functions, not shown here:
982
983
              def intersect_line_plane(self, P_line, P_plane)
984
              def intersect_two_planes(self, weights1, weights2)
          .....
985
```

```
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```

### A. Code Loading Edge Detection

```
986
          987
          # Alternative Approach
988
         def find_edge_points_centroids(self, pcd, radius=0.09, max_nn=40, tol=0.009):
989
990
991
             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.
992
993
             Problem of this method: Finds also points at the border of the point cloud.
994
             Parameters
 995
             pcd : open3d PointCloud
996
997
                 Point cloud, where the edge points should be found.
998
             radius : float, optional
                Radius around a point, where the kd-tree looks for neighbors. The default is 0.09.
999
1000
             max_nn : int, optional
1001
                 Maximum number of neighbors to be found. The default is 40.
1002
             tol : float, optional
                 Tolerance of the distance between the centroid and the point. The default is 0.009.
1003
                 If the distance is larger than tol, the point is considered an edge point.
1004
1005
             Returns
1006
1007
             edge_points_ind : list of int
1008
                 Indices of the edge points.
             .....
1009
1010
             # compute the kd-tree of the point cloud to enable fast neighbor search
1011
             kdtree = o3d.geometry.KDTreeFlann(pcd)
             # downsample the point cloud to speed up computation
1012
             downsampled_pcd = pcd.uniform_down_sample(self.uniform_downsample_factor)
1013
1014
             edge_points_ind = []
             for i in range(len(downsampled_pcd.points)):
1015
1016
                 # find the neighbors of the point
1017
                 _, point_indices, _ = kdtree.search_hybrid_vector_3d(
1018
                     query=downsampled_pcd.points[i], radius=radius, max_nn=max_nn
                 )
1019
1020
                 # get the point cloud of the neighbors
                 neighbor_pcd = pcd.select_by_index(point_indices)
1021
1022
                 # compute the centroid of the neighbors
                 sum_of_points = np.sum(np.asarray(neighbor_pcd.points), axis=0)
1023
                 centroid = sum_of_points / len(neighbor_pcd.points)
1024
                 # compute the distance between the centroid and the point
1025
1026
                 dist_to_point = np.linalg.norm(centroid - downsampled_pcd.points[i])
1027
                 # if the distance is larger than tol, the point is considered an edge point
1028
                 if dist_to_point > tol:
1029
                     edge_points_ind.append(self.uniform_downsample_factor * i)
1030
             return edge_points_ind
```

```
import gtsam
1
    import numpy as np
2
3
    import copy
    import pickle
4
    class Truck3D:
5
        def __init__(
6
             self.
7
            observation_noise_translation_default=0.17,
8
             observation_noise_rotation_default=0.14,
9
            construction_noise_translation_default=12,
10
11
             construction_noise_rotation_default=1.5,
            updating_factor_noise=0.94,
12
            updating_factor_weights=0.97
13
14
             initial_weights_translation=0.8,
15
             initial_weights_rotation=0.8,
16
            align_every_n_steps=32,
17
             saving_old_factors_until_n_time_steps=50,
             **truck_configuration,
18
19
        ):
             .....
20
21
            Constructor.
22
            Parameters
23
             observation_noise_translation_default : float, optional
24
                 Standard deviation of the noise for the translation of the observations.
25
                 The default is 0.17.
26
             observation\_noise\_rotation\_default\ :\ float,\ optional
27
                 Standard deviation of the noise for the rotation of the observations. The default is 0.14.
28
29
             construction_noise_translation_default : float, optional
                 Default standard deviation of the noise for the translation of the construction factors.
30
31
                 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.
32
             construction_noise_rotation_default : float, optional
33
                 Default standard deviation of the noise for the rotation of the construction factors.
34
35
                 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.
36
37
             updating_factor_noise : float, optional
                 Factor to update the construction noise after each time step.
38
39
                 Update the default construction noise with this factor after each time step.
                 The default is 0.94.
40
41
             updating_factor_weights : float, optional
                 Factor to update the weights of the factors after each time step.
42
                 The default is 0.97.
43
             initial_weights_translation : float, optional
44
                 Initial weight for updating the translation of parts of the truck after a new observation.
45
                 The default is 0.8.
46
             initial_weights_rotation : float, optional
47
                 Initial weight for updating the rotation of parts of the truck after a new observation.
48
49
                 The default is 0.8.
             align\_every\_n\_steps : int, optional
50
                 How often dependent parts of the model are aligned.
51
```

52	If 0, dependent parts are never aligned.
53	If 1, dependent parts are aligned after each time step.
54	The default is 32.
55	saving_old_factors_until_n_time_steps : int, optional
56	Number of time steps, old observation factors should stay in the graph.
57	The default is 50.
58	**truck_configuration : dict
59	Dictionary with the configuration of the truck. The keys of the dictionary are some
60	features of the truck (e.g. width of the loading platform). Not all features
61	have to be given (for the features that are not given, default values are used).
62	Attributes
63	
64	graph : gtsam.NonlinearFactorGraph
65	Current factor graph.
66	updating_factor_noise : float in (0, 1]
67	Factor to update the construction noise after each time step.
68	The construction noise is updated with the function update_construction_noise().
69	updating_factor_weights : float in (0, 1]
70	Factor to update the weights for model updates after each time step.
71	The weights are updated with the function update_weights().
72	lower_oouna_construction_noise : jloat >= 0
73	Lower bound for the construction noise in the updating function.
74	lower_bound_weights : float in [0, 1]
75	Lower oouna for the weights in the upating function.
76	werght_rotation : float in (lower_oound_werghts, 1)
77	current weight for the upactes of the rotation.
78	werght_translation : jioat in (lower_oowna_werghts, j
79	converting period potential defaults of the translation.
80	$s_{total down down to make the second down the second down to the second down the second down the second to the $
80	$\beta$ contains a dependence of the noise for the rotation of the cost outlons.
82	$s_{1} = s_{1} + s_{2} + s_{3} + s_{3$
84	construction noise rotation default · float
85	Default standard deviation of the noise for the rotation of the construction factors.
86	Used for construction factors of narts that are in no special relation to each other.
87	construction noise rotation dependent : float
88	Standard deviation of the noise for the rotation of the construction factors between
89	dependent parts.
90	construction_noise_translation_default : float
91	Default standard deviation of the noise for the translation of the construction factors.
92	Used for construction factors of parts that are in no special relation to each other.
93	construction_noise_translation_dependent : float
94	Standard deviation of the noise for the translation of the construction factors between
95	dependent parts.
96	align_every_n_steps : int >= 0
97	How often dependent parts of the model are aligned.
98	If 0, dependent parts are never aligned.
99	If 1, dependent parts are aligned after each time step.
100	factors_to_reuse : list of gtsam.NonlinearFactorGraph
101	List of the observation factors that are saved to reuse them after each time step
102	in the recreation of the factor graph. Old observations stay the same each time
103	the factor graph is recreated and should be kept in the factor graph for some time.
104	saving_old_factors_until_n_time_steps : int > 0
105	Number of time steps, old observation factors should stay in the graph.
106	time_step_factors_to_reuse : list of int
107	List of the corresponding time step of the factors to reuse.
108	Used to check if observations are too old and therefore deleted from the list.
109	symbols_sensor : list of gtsam.Symbol
110	List of the symbols for the sensor in each time step.
111	sensor_trajectory : list of gtsam.Pose3
112	List of the estimated poses of the sensor in the world frame.
113	current_num_time_steps : int

```
Number of time steps that have already been processed.
        114
        115
                       g0, G0, w1, w2, w3, W1, W2, W3, l1, l2, L1, L2 : gtsam.Symbol
        116
                           Symbols for the variables.
                           g0 ... origin of the world/global frame
        117
                           GO ... not any origin, but to stay consistent with the naming convention
        118
                           w1, w2, w3 ... left wheels (one front wheel (w1) and two rear wheels (w2, w3))
        119
        120
                           W1, W2, W3 ... right wheels (one front wheel (W1) and two rear wheels (W2, W3))
                           11, 12 ... left lights (one front light (11) and one rear light (12))
        121
                           L1, L2 ... right lights (one front light (L1) and one rear light (L2))
        122
        123
                       symbols\_parts : list of gtsam.Symbol
                           List of all symbols for the variables.
        124
        125
                       dependency_groups : dict
                           Dependency groups contain parts of the truck that have a fixed spatial relation
        126
                           to each other, e.g. all wheels have the same z-coordinate in the world frame.
        127
        128
                           Therefore, if the z-coordinate of one wheel is changed, the z-coordinates of all other
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        129
                           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.
        130
                       groups_with_degrees_of_freedom : dict
        131
        132
                           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
        133
                           a dictionary with the axes as keys. The values are again dictionarys containing
        134
                           groups of symbols that are in a fixed relation to each other. A group is given as
        135
                           a dictionary containing "symbols" (list of symbols) and "std" (additional
        136
        137
                           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).
        138
                       symbols_info : dict
        139
        140
                           Dictionary with information about each symbol defining the model of the truck implicitly.
        141
                           For each symbol, the following information is contained:
        142
                               num_observed: integer, indicating how often the symbol has been observed.
                               pose_in_world_frame: dictionary with keys "x", "y", "z", "rotation_matrix".
        143
                               dependency_groups: dictionary with keys "x", "y", "z", "roll", "pitch", "yaw"
        144
        145
                                    and the symbols in their respective group as values.
                               groups_with_degrees_of_freedom: dictionary with only the necessary keys of "x", "y",
        146
                                "z", "roll", "pitch", and "yaw" and the symbols in their respective group as values.
        147
        148
                               std_degrees_of_freedom: dictionary with the additional standard deviation for
                                    the respective groups with degrees of freedom.
        149
                       construction_noise_dictionary : dict
        150
                           Dictionary with the standard deviations of the noise models for the construction factors.
        151
        152
                           This information is saved such that it does not have to be computed in each time step.
        153
                       length_loading_platform : float
        154
                           Length of the loading platform.
        155
                       true_truck_configuration : dict
                           Dictionary that contains the ground truth of the truck configuration. Used for testing.
        156
                       .....
        157
                       self.graph = gtsam.NonlinearFactorGraph()
        158
                       \ensuremath{\texttt{\#}} define input parameters for noise, weights, and updating factors as attributes
        159
        160
                       self.updating_factor_noise = updating_factor_noise
                       self.updating_factor_weights = updating_factor_weights
        161
        162
                       self.lower_bound_construction_noise = 0.1
        163
                       self.lower_bound_weights = 0.1
                       self.weight_rotation = initial_weights_rotation
        164
                       self.weight_translation = initial_weights_translation
        165
        166
                       self.observation_noise_rotation_default = observation_noise_rotation_default
                       self.observation_noise_translation_default = observation_noise_translation_default
        167
        168
                       self.construction_noise_rotation_default = construction_noise_rotation_default
                       self.construction noise rotation dependent = 0
        169
        170
                       self.construction_noise_translation_default = construction_noise_translation_default
        171
                       self.construction_noise_translation_dependent = 0
        172
                       # define how often dependent parts of the model are aligned
                       self.align_every_n_steps = align_every_n_steps
        173
```

```
# define empty lists for recreating the factor graph
174
175
              self.factors_to_reuse = []
              self.saving_old_factors_until_n_time_steps = saving_old_factors_until_n_time_steps
176
177
              self.time_step_factors_to_reuse = []
              # define empty lists for the symbols and the estimated poses of the sensor
178
179
              self.symbols_sensor = []
180
              self.sensor_trajectory = []
181
              # define a counter for the number of time steps
              self.current_num_time_steps = 0
182
              # define all available symbols for the variables
183
              (
184
185
                  self.g0,
186
                  self.G0,
                  self.w1.
187
188
                  self.w2,
189
                  self.w3,
                  self.W1.
190
                  self.W2,
191
192
                  self.W3,
193
                  self.l1,
                  self.12,
194
                  self.L1.
195
196
                  self.L2,
              ) = self.define_symbols()
197
              self.symbols_parts = [
198
199
                  self.g0,
                  self.GO,
200
201
                  self.w1,
                  self.w2,
202
203
                  self.w3.
204
                  self.W1,
205
                  self.W2,
                  self.W3.
206
207
                  self.l1,
208
                  self.12,
                  self.L1,
209
210
                  self.L2,
              ]
211
212
              # dictionary with symbols that are spatially dependent on each other in some way (i.e. axis)
              self.dependency_groups = {
213
214
                  "x": {
                      0: [self.g0, self.G0],
215
                      1: [self.w1, self.W1],
216
217
                      2: [self.w2, self.W2],
                      3: [self.w3, self.W3],
218
219
                      4: [self.l1, self.L1],
                  },
220
                   "y": {
221
                      0: [self.w1, self.w3],
222
                      1: [self.W1, self.W3],
223
                  },
224
                  "z": {
225
                      0: [self.w1, self.w2, self.w3, self.W1, self.W2, self.W3],
226
                      1: [self.l1, self.L1],
227
228
                      2: [self.12, self.L2],
                  },
229
              }
230
231
              # dictionary with symbols that have some degrees of freedom in their relation to other symbols
```

```
232
             # and the parts that are in a fixed relation to them
233
             # the additional uncertainty is given as the standard deviation of the noise model ("std")
234
             self.groups_with_degrees_of_freedom = {
                 "yaw": {
235
                     # the yaw of the front wheels might change depending on the steering angle
236
                    0: {
237
                         "symbols": [self.w1, self.W1],
238
                        "std": 1.5 * np.pi,
239
                        "rotate_second_matrix": np.pi,
240
                    },
241
242
                },
                 "z": {
243
                     # the height of the loading edge might change depending on the load
244
                    0: {"symbols": [self.g0, self.G0], "std": 2},
245
                },
246
247
             }
248
             \ensuremath{\texttt{\#}} dictionary with information about the variables
             self.symbols_info = {}
249
             self.build_symbols_info()
250
             # get the construction noise model depending on the degrees of freedom and dependent parts
251
252
             self.construction_noise_dictionary = {}
253
             self.build_construction_noise_dictionary()
254
             # define the approximate truck configuration
             self.length_loading_platform = 5.87
255
             self.define_estimated_truck_configuration(truck_configuration)
256
             # define the ground truth configuration of the truck
257
258
             self.true_truck_configuration = {}
259
             self.define_ground_truth_truck_configuration()
         260
261
         # Definition of the truck
         *****
262
263
         def build_symbols_info(self):
264
             Build the symbols_info dictionary at the initialization of the class.
265
                 It contains for each symbol the number of times this symbol has been observed,
266
267
                 the poses of the parts in the world frame,
                 the dependency groups the symbol is involved in,
268
                 and the groups with degrees of freedom the symbol is involved in
269
270
                 with the respective standard deviation.
             .....
271
             self.symbols_info = {
272
                 symbol: {
273
                     "num_observed": 0,
274
275
                     "pose_in_world_frame": {
                        "x": 0,
276
                        "y": 0,
277
                        "z": 0,
278
279
                        "rotation_matrix": gtsam.Rot3(),
                    }.
280
                     "dependency_groups": {
281
                         "x": [],
282
                         "y": [],
283
                        "z": [],
284
                        "roll": [],
285
286
                         "pitch": [],
287
                        "yaw": [],
                    },
288
```

```
"groups_with_degrees_of_freedom": {},
289
                       "std_degrees_of_freedom": {},
290
                  }
291
292
                  for symbol in self.symbols_parts
              }
293
294
              # adapt the dependency groups and groups with degrees of freedom for each symbol
295
              for symbol in self.symbols_parts:
                  for axis in self.dependency_groups:
296
297
                       for group in self.dependency_groups[axis]:
                           # if the symbol is in this specific group, we add all other symbols in this group
298
299
                           # to the dependency groups of the symbol
300
                           if symbol in self.dependency_groups[axis][group]:
                               for other_symbol in self.dependency_groups[axis][group]:
301
302
                                   if other_symbol != symbol:
303
                                       self.symbols_info[symbol]["dependency_groups"][axis].append(
304
                                            other_symbol
                                       )
305
306
                  for axis in self.groups_with_degrees_of_freedom:
307
                       for group in self.groups_with_degrees_of_freedom[axis]:
                           if symbol in self.groups_with_degrees_of_freedom[axis][group]["symbols"]:
308
309
                               # 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
310
                               \ensuremath{\texttt{\#}} deviation of the degrees of freedom to the symbols_info dictionary
311
312
                               # gather all other symbols in this group
313
                               other symbols = []
314
                               for other_symbol in self.groups_with_degrees_of_freedom[axis][group][
                                    "symbols"
315
316
                               1:
317
                                    if other_symbol != symbol:
                                       other_symbols.append(other_symbol)
318
319
                               self.symbols_info[symbol] ["groups_with_degrees_of_freedom"] [
320
                                   axis
                               ] = other_symbols
321
                               self.symbols_info[symbol]["std_degrees_of_freedom"][
322
323
                                   axis
324
                               ] = self.groups_with_degrees_of_freedom[axis][group]["std"]
325
          def define_estimated_truck_configuration(self, truck_configuration={}):
326
327
              Define the estimated configuration of the truck by defining the poses of all parts
              in the world frame in the symbols_info dictionary.
328
              Calls the function define_symbols_info_for_estimated_truck_configuration().
329
330
              Parameters
331
332
              truck_configuration : dict, optional
333
                  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
334
                  given (for the features that are not given, default values are used). The default is {}.
335
              .....
336
              \ensuremath{\textit{\#}} we have to check whether the length of the loading platform is given
337
              if "length_loading_platform" not in truck_configuration:
338
339
                  self.length_loading_platform = 5.87
340
              else:
                  self.length_loading_platform = truck_configuration["length_loading_platform"]
341
              # define the poses of all parts in the world frame in the symbols_info dictionary
342
              # pass the truck configuration with ''**'' to unpack the dictionary
343
344
              self.define_symbols_info_for_estimated_truck_configuration(**truck_configuration)
```

```
345
                  346
                  347
                  348
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                  358
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                  403
                  404
                  405
```

```
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_g0_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:
```

def define\_symbols\_info\_for\_estimated\_truck\_configuration(

self.

```
rotation_matrix_front_lights = self.get_rotation_matrix(np.pi / 2, axis="z")
406
              if rotation_matrix_rear_lights is None:
407
408
                  rotation_matrix_rear_lights = self.get_rotation_matrix(-np.pi / 2, axis="z")
              y_middle = width_loading_platform / 2
409
              # define the poses of all parts in the world frame in the symbols_info dictionary
410
411
              self.symbols_info[self.g0]["pose_in_world_frame"] = {
                  "x": 0.
412
413
                  "y": 0,
                  "z": 0,
414
                  "rotation_matrix": gtsam.Rot3(),
415
              7
416
              self.symbols_info[self.G0]["pose_in_world_frame"] = {
417
418
                  "x": 0,
419
                  "y": width_loading_platform,
                  "z": 0,
420
421
                  "rotation_matrix": gtsam.Rot3(rotation_matrix_right_loading_edge),
              }
422
              self.symbols_info[self.w1]["pose_in_world_frame"] = {
423
424
                  "x": -distance_g0_w1,
                  "y": 0,
425
                  "z": -height_loading_platform_above_wheels,
426
427
                  "rotation_matrix": gtsam.Rot3(),
              }
428
429
              self.symbols_info[self.W1]["pose_in_world_frame"] = {
430
                  "x": -distance_g0_w1,
                  "y": width_loading_platform,
431
432
                  "z": -height_loading_platform_above_wheels,
                  "rotation_matrix": gtsam.Rot3(rotation_matrix_right_wheels),
433
434
              }
435
              self.symbols_info[self.w2]["pose_in_world_frame"] = {
                  "x": -distance_g0_w1 + distance_w1_w2,
436
                  "y": 0,
437
438
                  "z": -height_loading_platform_above_wheels,
                  "rotation_matrix": gtsam.Rot3(),
439
              }
440
              self.symbols_info[self.W2]["pose_in_world_frame"] = {
441
442
                  "x": -distance_g0_w1 + distance_w1_w2,
443
                  "y": width_loading_platform,
                  "z": -height_loading_platform_above_wheels,
444
445
                  "rotation_matrix": gtsam.Rot3(rotation_matrix_right_wheels),
              }
446
447
              self.symbols_info[self.w3]["pose_in_world_frame"] = {
                  "x": -distance_g0_w1 + distance_w1_w2 + distance_w2_w3,
448
                  "γ": 0,
449
                  "z": -height_loading_platform_above_wheels,
450
451
                  "rotation_matrix": gtsam.Rot3(),
              }
452
453
              self.symbols_info[self.W3]["pose_in_world_frame"] = {
454
                  "x": -distance_g0_w1 + distance_w1_w2 + distance_w2_w3,
                  "v": width_loading_platform,
455
                  "z": -height_loading_platform_above_wheels,
456
457
                  "rotation_matrix": gtsam.Rot3(rotation_matrix_right_wheels),
              }
458
              self.symbols_info[self.l1]["pose_in_world_frame"] = {
459
                  "x": -distance_lights_in_front_of_loading_platform,
460
                  "y": y_middle - distance_front_lights / 2;
461
                  "z": -height_loading_platform_above_front_lights,
462
                  "rotation_matrix": gtsam.Rot3(rotation_matrix_front_lights),
463
              }
464
              self.symbols_info[self.L1]["pose_in_world_frame"] = {
465
                  "x": -distance_lights_in_front_of_loading_platform,
466
```

```
"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.12]["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 around 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
       \ensuremath{\texttt{\#}} 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())
```

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515

516

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518

519

```
for symbol in self.symbols_info.keys():
521
                  new_pose = self.symbols_info_to_pose_in_world_frame(symbol).compose(trafo_g0_to_origin)
522
                  self.symbols_info[symbol]["pose_in_world_frame"] = {
523
                      "x": new_pose.translation()[0],
524
                      "y": new_pose.translation()[1],
525
                      "z": new_pose.translation()[2],
526
527
                      "rotation_matrix": new_pose.rotation(),
                  7
528
529
          def update_pose_in_world_frame(self, symbol, new_pose):
530
              Update the pose of a part in the world frame according to the updating weights.
531
532
              Parameters
533
              symbol : gtsam.Symbol
534
535
                  Symbol of the part.
536
              new_pose : gtsam.Pose3
537
                  New estimated pose.
              .....
538
              old_pose = self.symbols_info_to_pose_in_world_frame(symbol)
539
              # update the rotation and translation according to the weights
540
              updated_rotation = self.update_rotation(old_pose, new_pose, self.weight_rotation)
541
542
              updated_translation = self.update_translation(old_pose, new_pose, self.weight_translation)
              # update the pose in the symbols_info dictionary
543
544
              self.symbols_info[symbol]["pose_in_world_frame"] = {
                  "x": updated_translation[0],
545
                  "y": updated_translation[1],
546
547
                  "z": updated_translation[2],
                  "rotation_matrix": updated_rotation,
548
              }
549
          def update_rotation(self, old_pose, new_pose, weight):
550
551
              Update the rotation of a pose according to some weight.
552
553
              Parameters
554
555
              old_pose : gtsam.Pose3
556
                 Old pose.
557
              new_pose : gtsam.Pose3
558
                  New pose.
              weight : float
559
560
                  Weight for the update.
              Returns
561
562
              qtsam.Rot3
563
                  Updated rotation.
564
              .....
565
566
              rotation_matrix_new = new_pose.rotation().matrix()
              rotation_matrix_old = old_pose.rotation().matrix()
567
              # compute the naive weighted average
568
              average_rotation_matrix = weight * rotation_matrix_new + (1 - weight) * rotation_matrix_old
569
              # project this matrix onto SO(3)
570
571
              # compute the singular value decomposition of the matrix
572
              U, _, V_t = np.linalg.svd(average_rotation_matrix, full_matrices=False)
573
              S = np.diag([1, 1, np.sign(np.linalg.det(U) * np.linalg.det(V_t))])
```

```
average_rotation_matrix = U @ S @ V_t
574
575
              return gtsam.Rot3(average_rotation_matrix)
          def update_translation(self, old_pose, new_pose, weight):
576
577
578
              Update the translation of a pose according to some weight.
             Parameters
579
580
              old_pose : gtsam.Pose3
581
582
                  Old pose.
583
              new_pose : gtsam.Pose3
584
                 New pose.
585
              weight : float
586
                  Weight of the new pose.
587
             Returns
588
              atsam.Point3
589
590
                 Updated translation.
              .....
591
             return weight * new_pose.translation() + (1 - weight) * old_pose.translation()
592
          def update_dependent_parameters(self):
593
594
595
             After changing the configuration of the truck by changing the poses of some parts
596
              in the world frame, we have to update the parameters dependent on these poses,
597
              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.
598
599
600
              weight_g0 = max(self.symbols_info[self.g0]["num_observed"], 1)
             weight_GO = max(self.symbols_info[self.GO]["num_observed"], 1)
601
602
             weight_12 = max(self.symbols_info[self.12]["num_observed"], 1)
603
             weight_L2 = max(self.symbols_info[self.L2]["num_observed"], 1)
604
             x_front_end = (
                  weight_g0 * self.symbols_info[self.g0]["pose_in_world_frame"]["x"]
605
                  + weight_G0 * self.symbols_info[self.G0]["pose_in_world_frame"]["x"]
606
607
             ) / (weight_g0 + weight_G0)
608
             x rear end = (
                  weight_12 * self.symbols_info[self.12]["pose_in_world_frame"]["x"]
609
                  + weight_L2 * self.symbols_info[self.L2]["pose_in_world_frame"]["x"]
610
611
             ) / (weight_12 + weight_L2)
612
             distance_lights_behind_loading_platform = 1.04
613
             self.length_loading_platform = (
614
                  abs(x_rear_end - x_front_end) - distance_lights_behind_loading_platform
              )
615
          def align_dependent_parts(self):
616
617
              If the position of a part is updated, the positions of the dependent parts have to be updated
618
619
              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.
620
              ......
621
              # we iterate over all axes ...
622
623
             for axis in self.dependency_groups.keys():
                  # ... and all groups to calculate the mean of the updated symbols in the current group
624
                  for group_list in self.dependency_groups[axis].values():
625
                        average for translation axes
626
                      if axis in ["x", "y", "z"]:
627
                          average_translation = self.get_average_translation(group_list, axis)
628
```

```
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```

629		for symbol in group_list:
630		<pre>self.symbols_info[symbol]["pose_in_world_frame"][axis] = average_translation</pre>
631		# average for rotation axes
632		else:
633		<pre>average_rotation_matrix, _ = self.get_average_rotation(group_list)</pre>
634		for symbol in group_list:
635		self.symbols_info[symbol]["pose_in_world_frame"][
636		"rotation_matrix"
637		] = gtsam.Rot3(average_rotation_matrix)
638		# and do the same for the aroups with dearees of freedom
639		for axis in self groups with degrees of freedom.keys():
640		for group in self.groups_with_degrees_of_freedom[axis].values():
641		group_list = group["symbols"]
642		if axis in ["x", "y", "z"]:
643		average translation = self.get average translation(group list, axis)
644		for symbol in group_list:
645		<pre>self.symbols_info[symbol]["pose_in_world_frame"][axis] = average_translation</pre>
646		else:
647		# we have to check if the rotation should be the same or rotated by a fixed angle
648		average_rotation_matrix1, average_rotation_matrix2 = self.get_average_rotation(
649		group_list, axis, rotate_second_matrix=group["rotate_second_matrix"]
650		)
651		<pre>self.symbols_info[group_list[0]]["pose_in_world_frame"][</pre>
652		"rotation_matrix"
653		] = gtsam.Rot3(average_rotation_matrix1)
654		<pre>self.symbols_info[group_list[1]]["pose_in_world_frame"][</pre>
655		"rotation_matrix"
656		] = gtsam.Rot3(average_rotation_matrix2)
657	dof	not average translation (self list of symbols avis).
007	der	get_average_transfatton(sett, fist_of_symbols, axis).
650		Cet the average translation of a list of symbols in a certain direction
660		The poses are weighted according to how often they have been observed.
661		Parameters
662		
663		list of symbols : list of atsam.Symbol
664		The list of symbols, we want to find the average of.
665		axis : str
666		The axis for which we want to find the average. One of $["x", "y", "z"]$ .
667		Returns
668		
669		weighted_pose_average : float
670		The average translation in the given direction.
671		
672		<pre>sum_pose, sum_observed = 0, 0</pre>
673		for symbol in list_of_symbols:
674		<pre>obs = max(self.symbols_info[symbol]["num_observed"], 1)</pre>
675		<pre>sum_pose += self.symbols_info[symbol]["pose_in_world_frame"][axis] * obs</pre>
676		<pre>sum_observed += obs</pre>
677		<pre>weighted_pose_average = sum_pose / sum_observed</pre>
678		return weighted_pose_average
679	def	<pre>get_average_rotation(self, list_of_symbols, axis=None. rotate second matrix=False):</pre>
680		

```
Get the average rotation of a list of rotation matrices. If the rotations are dependent on
681
682
             each other, but rotated by a fixed angle, one matrix is rotated by this angle,
683
            then they are averaged, and then this matrix is rotated back.
684
            Parameters
685
686
             list_of_symbols : list of gtsam.Symbol
687
                The list of symbols, we want to find the average of.
            axis : str
688
689
                The axis around which we want to find the average. One of ["roll", "pitch", "yaw"].
690
            rotate_second_matrix : bool or angle, optional
                If not False, this is the angle around which the second rotation matrix is rotated.
691
692
            Returns
693
694
            average_rotation_matrix : ndarray
                The average rotation around the given axis.
695
696
697
            if rotate_second_matrix is not False:
                rot_matrix1 = self.symbols_info[list_of_symbols[0]]["pose_in_world_frame"][
698
699
                    "rotation_matrix"
700
                l.matrix()
                rot_matrix2 = self.symbols_info[list_of_symbols[1]]["pose_in_world_frame"][
701
702
                    "rotation_matrix"
                l.matrix()
703
704
                angle = rotate_second_matrix
705
                helper_rot_matrix = self.get_rotation_matrix(angle, axis)
706
                rot_matrix2 = helper_rot_matrix @ rot_matrix2
707
                # compute the naive average
                average_rotation_matrix = (rot_matrix1 + rot_matrix2) / 2
708
709
            else:
710
                list_of_matrices = []
                for symbol in list_of_symbols:
711
                    list_of_matrices.append(
712
                        self.symbols_info[symbol]["pose_in_world_frame"]["rotation_matrix"].matrix()
713
                    )
714
715
                # compute the naive average
716
                average_rotation_matrix = np.sum(np.array(list_of_matrices), axis=0) / len(
717
                    list_of_matrices
                )
718
            # project this matrix onto SO(3) and compute the singular value decomposition
719
720
            U, _, V_t = np.linalg.svd(average_rotation_matrix, full_matrices=False)
721
            S = np.diag([1, 1, np.sign(np.linalg.det(U) * np.linalg.det(V_t))])
            average_rotation_matrix = U @ S @ V_t
722
723
            if rotate_second_matrix:
                return average_rotation_matrix, helper_rot_matrix.T @ average_rotation_matrix
724
            else:
725
726
                return average_rotation_matrix, None
         727
         # Pose estimation
728
         729
        def estimate_sensor_pose(self, observations, observations_info=None, random_guess=False):
730
731
            Estimate the pose of the sensor in the world frame that fits the observations best.
732
```

733	Parameters
734	
735	ooservations : atcl
736	Dictionary with the symbols of the observed parts as keys and the observations as values.
737	observations_injo : dict, optional
738	Dictionary with the symbols of the observed parts as keys and the information about the
739	observations as values. Such information is an axis with certain degree of freedom due to
740	incomplete observations, e.g. we only observe a part of the loading edge
741	but not its beginning. The default is None.
742	random_guess : bool, optional
743	If True or the sensor trajectory is empty, the initial guess for the optimization
744	of the sensor pose is a random pose.
745	Else, the initial guess for the optimization is the estimated pose of the sensor
746	in the previous time step.
747	The default is False.
748	Returns
749	
750	best_pose_new_sensor : gtsam.Pose3
751	Estimated pose of the sensor in the world frame.
752	best_error : float
753	Error of the optimization. Note: This is not the error of the estimated sensor pose
754	to the ground truth, but the error of the optimization.
755	result : gtsam.Values
756	Result of the optimization.
757	
758	# define initial guess for the optimization
759	if random_guess or self.current_num_time_steps == 0:
760	initial_guess_sensor_pose = gtsam.Pose3(
761	<pre>gtsam.Rot3(), gtsam.Point3(np.random.rand(3) * 10)</pre>
762	)
763	else:
764	<pre>initial_guess_sensor_pose = self.get_current_sensor_pose()</pre>
765	# insert the observations as factors in the factor granh
766	self.observe(observations, observations_info)
767	# get the initial values for all parts and previous sensor poses
768	<pre>initial = self.get_initial_values()</pre>
769	# estimate the pose of the sensor
770	<pre>best_pose_new_sensor, best_error, result = self.optimize_sensor_pose(</pre>
771	initial, initial_guess_sensor_pose
772	)
773	# add the estimated pose to the sensor trajectory and the symbols_info dictionary
774	<pre>self.add_pose_to_trajectory(best_pose_new_sensor)</pre>
775	return best_pose_new_sensor, best_error, result
776 def	<pre>observe(self, observations, observations_info=None):</pre>
777	"""
778	First, we recreate the factor graph with the new estimated truck configuration.
779	Then, we add a new sensor to the factor graph.
780	Finally, add the observations as factors to the factor graph.
781	Parameters
782	
783	observations : dict
784	Dictionary with the symbols of the observed parts as keys and the observations as values.
785	observations_info : dict, optional
786	Dictionary with the symbols of the observed parts as keys and the information about the
787	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
788
789
                  but not its beginning. The default is None.
              .....
790
              # create the factor graph from scratch (this substitutes the old factor graph)
791
792
              self.create_factor_graph()
              # add the old observation factors
793
              indices_to_delete = []
794
              for i in range(len(self.factors_to_reuse)):
795
796
                  if (
797
                      self.time_step_factors_to_reuse[i]
                      < self.current_num_time_steps - self.saving_old_factors_until_n_time_steps
798
                  ):
799
                      indices_to_delete.append(i)
800
801
                  # the list is monotonically increasing, so we can break here
                  else:
802
803
                      break
              # delete the old factors from the list of factors to reuse
804
              for i in reversed(indices_to_delete):
805
                  del self.factors_to_reuse[i]
806
                  del self.time_step_factors_to_reuse[i]
807
              # add the old factors to the new factor graph
808
809
              for factor in self.factors_to_reuse:
810
                  self.graph.add(factor)
811
              # 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:
812
813
                  self.symbols_info.pop(
814
                      self.get_sensor_symbol(
                          self.current_num_time_steps - self.saving_old_factors_until_n_time_steps - 1
815
                      )
816
817
                  )
              # add a new sensor and the observations as factors to the factor graph
818
              self.add_new_sensor(observations, observations_info)
819
820
          def add_new_sensor(self, observations, observations_info):
821
822
              Add a new sensor and the observations to the factor graph of the current time step.
823
              Parameters
824
              observations : dict
825
                  Dictionary with the symbols of the observed parts as keys and the observations as values.
826
827
              observations_info : dict, optional
                  Dictionary with the symbols of the observed parts as keys and the information about the
828
829
                  observations as values. Such information is an axis with certain degree of freedom due to
830
                  incomplete observations, e.g. we only observe a part of the loading edge
                  but not its beginning. The default is None.
831
              .....
832
833
              # get the symbol for the sensor in the next time step and add it to the list of sensor symbols
834
              next_time_step = self.current_num_time_steps
              symbol_new_sensor = self.get_sensor_symbol(next_time_step)
835
              self.symbols_sensor.append(symbol_new_sensor)
836
              # add the observations as factors to the factor graph
837
              self.observations_to_factors(symbol_new_sensor, observations, observations_info)
838
839
          def observations_to_factors(self, sensor_symbol, observations, observations_info):
840
```

841	Convert the observations to factors between the sensor and the parts and insert them into the
842	factor graph. Used in the function add_new_sensor(). The observations_info dictionary gives
843	additional information about the observations, i.e. uncertainty in certain directions,
844	e.g. observations_info = {w1: {"x": 2.4, "roll": 0.6*np.pi}}
845	$\ldots$ if the observation of w1 is unsure in x-direction and roll rotation.
846	Parameters
847	
848	sensor_symbol : gtsam.Symbol
849	Symbol of the sensor in the current time step.
850	observations : dict
851	Dictionary with the symbols of the observed parts as keys and the observations as values.
852	observations_info : dict
853	Dictionary with the symbols of the observed parts as keys and the information about the
854	observations as values. Such information is an axis with certain degree of freedom due to
855	incomplete observations, e.g. we only observe a part of the loading edge
856	but not its beginning. The default is None.
857	"""
858	for symbol in observations:
859	# define the observation noise model
860	observation_noise_model = {
861	"roll": self.observation_noise_rotation_default,
862	" <pre>pitch": self.observation_noise_rotation_default,</pre>
863	"yaw": self.observation_noise_rotation_default,
864	"x": self.observation_noise_translation_default,
865	"y": self.observation_noise_translation_default,
866	"z": self.observation_noise_translation_default,
867	}
868	# adapt the observation noise model according to observations_info
869	if observations_info is not None:
870	if symbol in observations_info.keys():
871	for axis in observations_info[symbol].keys():
872	<pre>observation_noise_model[axis] = observations_info[symbol][axis]</pre>
873	# define the observation noise model
874	<pre>observation_noise_model = self.get_noise_model(**observation_noise_model)</pre>
075	t define the factor and old it to the aranh
873	factor = fractor and add it to the graph
8/0	Iactor - gusam.betweenractorroses(
8//	Sensor_Symbol, Symbol, Observations[Symbol], Observation_noise_model
879	self granh add(factor)
015	borr. Bruphi add (rabbor)
880	# add the factor to the list of factors that should be reused in the next time step
881	<pre>self.factors_to_reuse.append(factor)</pre>
882	<pre>self.time_step_factors_to_reuse.append(self.current_num_time_steps)</pre>
883	# adapt the num observed attribute of the observed nart
884	<pre>self.symbols_info[symbol] ["num_observed"] += 1</pre>
885	# adapt the weights for updating rotation and translation
886	self.update_weights("translation")
887	self.update_weights("rotation")
888	<pre>def optimize_sensor_pose(self, initial, initial_guess_new_sensor):</pre>
889	""
890	Find the pose of the sensor in the world frame that fits the observations and the assumed
891	model best. The initial guess is the estimated pose of the sensor in the world frame at the
892	beginning of the optimization.
893	Parameters
894	

```
initial : gtsam. Values
895
896
                  Initial values of the variables except the new sensor.
897
              initial_guess_new_sensor : gtsam.Pose3
898
                  Initial guess for the pose of the sensor in the world frame.
899
              Returns
900
901
              best_pose_new_sensor : gtsam.Pose3
                  Pose of the sensor in the world frame that fits the observations best.
902
903
              best_error : float
904
                 Optimization error of the best pose of the sensor in the world frame.
905
              result : gtsam.Values
                  Optimized values of all variables.
906
              .....
907
908
              # get the current sensor symbol
909
              symbol_new_sensor = self.get_sensor_symbol(self.current_num_time_steps)
910
              \ensuremath{\textit{\#}} insert the initial guess into the current result
              initial.insert(symbol_new_sensor, initial_guess_new_sensor)
911
              # optimize the graph with the current result
912
              result = self.optimize_LM(initial)
913
              error = self.graph.error(result)
914
915
              new_pose_new_sensor = result.atPose3(symbol_new_sensor)
916
              return new_pose_new_sensor, error, result
          def add_pose_to_trajectory(self, pose):
917
918
              Add the estimated pose of the sensor in the world frame to the sensor trajectory and to the
919
920
              symbols_info dictionary, increase the number of time steps by one, and update the default
921
              construction noise.
922
              Parameters
923
              pose : gtsam.Pose3
924
                  Estimated pose of the sensor in the world frame.
925
              .....
926
              # add the pose to the sensor trajectory
927
928
              self.sensor_trajectory.append(pose)
929
              # add a new entry to the symbols_info dictionary for the sensor in the new time step
930
              self.symbols_info[self.get_sensor_symbol(self.current_num_time_steps)] = {
931
                  "num_observed": 0.
                  "pose_in_world_frame": {
932
                      "x": pose.translation()[0],
933
                      "y": pose.translation()[1],
934
935
                      "z": pose.translation()[2],
                      "rotation_matrix": pose.rotation(),
936
937
                  },
938
                  "dependency_groups": {},
                   "groups_with_degrees_of_freedom": {},
939
                  "std_degrees_of_freedom": {},
940
941
              }
              # increase the number of time steps by one
942
              self.current_num_time_steps += 1
943
              # adapt construction noise
944
              self.construction_noise_rotation_default *= self.updating_factor_noise
945
              self.construction_noise_translation_default *= self.updating_factor_noise
946
          .....
947
```

```
Additional functions, not shown here:
948
             def get_initial_values(self)
949
950
          951
          # Factor araph
952
         953
         def build_construction_noise_dictionary(self):
954
955
             Build the construction noise dictionary. This function is called once in the beginning such
956
957
             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
958
             construction noise for each axis,
959
             e.g. {"x": 0.1, "y": 0.1, "z": 0.1, "roll": 0.1, "pitch": 0.1, "yaw": 0.1}.
960
961
             # define which variables are connected by factors
962
             self.define_used_factors()
963
             # the order of the symbols stays the same, so we do not have to check each pair twice
964
             for tuple_of_symbols in self.construction_noise_dictionary.keys():
965
                 first_symbol = tuple_of_symbols[0]
966
                 second_symbol = tuple_of_symbols[1]
967
                 # initialize the construction noise with "default"
968
969
                 # this stays the same if the parts are not dependent on each other
970
                 construction_noise = {
                     "roll": "default".
971
972
                     "pitch": "default",
                     "yaw": "default",
973
                     "x": "default",
974
                     "y": "default"
975
                     "z": "default",
976
                 }
977
                 # adapt the construction noise according to the dependency groups
978
                 for axis in self.symbols_info[first_symbol]["dependency_groups"].keys():
979
                     if second_symbol in self.symbols_info[first_symbol]["dependency_groups"][axis]:
980
                         construction_noise[axis] = "dependent"
981
                 # adapt the construction noise according to the degrees of freedom
982
983
                 for axis in self.symbols_info[first_symbol]["groups_with_degrees_of_freedom"]:
984
                     new_construction_noise = self.get_construction_noise_with_degrees_of_freedom(
985
                         first_symbol, second_symbol, axis
                     )
986
                     construction_noise[axis] = new_construction_noise
987
988
                 # add the construction noise to the dictionary
                 self.construction_noise_dictionary[tuple_of_symbols] = copy.deepcopy(construction_noise)
989
         def define_used_factors(self):
990
991
             Define which parts of the truck should be connected with factors.
992
993
             The variables have a fixed order, so each combination of symbols appears exactly once.
             Define the construction noise dictionary.
994
995
             .....
             # define the keys in the form of a dictionary (if w1 is in factors[g0], [(g0, w1)] is a key)
996
997
             factors = {
                 self.g0: [self.l1, self.w1, self.w2, self.G0],
998
                 self.GO: [self.L1, self.W1, self.W2],
999
1000
                 self.w1: [self.l1, self.w2, self.W1],
1001
                 self.w2: [self.w3, self.W2],
1002
                 self.w3: [self.l2, self.W3],
```

```
1003
                  self.W1: [self.L1, self.W2],
1004
                   self.W2: [self.W3],
                   self.W3: [self.L2],
1005
1006
                   self.l1: [self.L1],
                   self.12: [self.L2],
1007
                   self.L1: [],
1008
1009
                   self.L2: [],
1010
              }
1011
               # build the structure of the construction noise dictionary
1012
               for symbol in factors.keys():
                   for other_symbol in factors[symbol]:
1013
                       self.construction_noise_dictionary[(symbol, other_symbol)] = {}
1014
1015
          def create_factor_graph(self):
1016
               (Re-)Create the factor graph and add the prior and construction factors to the graph.
1017
1018
               .....
1019
               # create an empty graph
              self.graph = gtsam.NonlinearFactorGraph()
1020
               # the world origin is exactly at the start of the left loading edge
1021
               # therefore, we set the respective noise to zero
1022
              world_origin_noise = gtsam.noiseModel.Diagonal.Sigmas(np.array([0] * 6))
1023
1024
               # define the prior factor and add it to the graph
1025
              factor_prior = gtsam.PriorFactorPose3(self.g0, gtsam.Pose3(), world_origin_noise)
              self.graph.add(factor_prior)
1026
1027
               # add the construction factors
1028
              self.add_construction_factors()
1029
          def add_construction_factors(self):
1030
1031
               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).
1032
1033
               The values of the BetweenFactorPose3 are based on the transformation between the poses of
1034
               two parts in the world frame (as stated in symbols_info). The noise model is based on
1035
               the construction noise dictionary.
1036
               .....
               # define the factors between all pairs of parts (order as stated in self.symbols_parts)
1037
1038
              for tuple_of_symbols in self.construction_noise_dictionary.keys():
                   first_symbol = tuple_of_symbols[0]
1039
                   second_symbol = tuple_of_symbols[1]
1040
1041
                   # get the transformation between the two parts
                   trafo_first_to_second_symbol = self.symbols_info_to_pose_in_world_frame(
1042
1043
                       first_symbol
                   ).between(self.symbols_info_to_pose_in_world_frame(second_symbol))
1044
                   construction_noise_model = copy.deepcopy(
1045
                       self.construction_noise_dictionary[tuple_of_symbols]
1046
                   )
1047
                   # substitute the entries in the construction noise model by their corresponding values
1048
1049
                   for axis in construction_noise_model.keys():
                       if construction_noise_model[axis] == "default":
1050
                           if axis in ["x", "y", "z"]:
1051
                               construction_noise_model[axis] = self.construction_noise_translation_default
1052
                           else:
1053
1054
                               construction_noise_model[axis] = self.construction_noise_rotation_default
                       elif construction_noise_model[axis] == "dependent":
1055
```

```
if axis in ["x", "y", "z"]:
1056
1057
                                                         construction_noise_model[axis] = self.construction_noise_translation_dependent
1058
                                                  else:
                                                         construction_noise_model[axis] = self.construction_noise_rotation_dependent
1059
                                          # if it is not "default" or "dependent", it gives the additional noise for this factor
1060
1061
                                          else:
                                                  if axis in ["x", "y", "z"]:
1062
                                                         construction_noise_model[axis] += self.construction_noise_translation_default
1063
1064
                                                  else:
                                                         construction_noise_model[axis] += self.construction_noise_rotation_default
1065
                                   # get the noise model for the factor
1066
                                  construction_noise_model = self.get_noise_model(**construction_noise_model)
1067
1068
                                   # define the factor and add it to the graph
                                  factor = gtsam.BetweenFactorPose3(
1069
1070
                                          first_symbol, second_symbol, trafo_first_to_second_symbol, construction_noise_model
                                  )
1071
1072
                                  self.graph.add(factor)
                   def update_construction_noise(self):
1073
1074
1075
                           Update the default construction noise for rotation and translation according to the number of
1076
                           steps. The construction noise follows the function % \left( f_{1}, f_{2}, f_{3}, 
1077
                                  c(n) = lambda^n * (c(0) - lower_bound) + lower_bound,
1078
                           thus we compute the new construction noise as
                                 c(n+1) = lambda * c(n) + (1-lambda)*lower_bound.
1079
1080
                           self.construction_noise_translation_default = (
1081
1082
                                   self.updating_factor_noise * self.construction_noise_translation_default
1083
                                   + (1 - self.updating_factor_noise) * self.lower_bound_construction_noise
1084
                           )
1085
                           self.construction_noise_rotation_default = (
1086
                                  self.updating_factor_noise * self.construction_noise_rotation_default
                                   + (1 - self.updating_factor_noise) * self.lower_bound_construction_noise
1087
                           )
1088
                   def update_weights(self, update_type):
1089
1090
1091
                           Update the weights of a part according to how often it has been observed.
1092
                           The weights follow the function
1093
                                  w(n) = lambda^n * (w(0) - lower_bound) + lower_bound,
                           thus we compute the new weight as
1094
1095
                                  w(n+1) = lambda * w(n) + (1-lambda)*lower_bound.
1096
                           Parameters
1097
                           update_type : str
1098
1099
                                  Either "translation" or "rotation".
1100
                           if update_type == "translation":
1101
                                  self.weight_translation = (
1102
1103
                                          self.updating_factor_weights * self.weight_translation
                                          + (1 - self.updating_factor_weights) * self.lower_bound_weights
1104
                                  )
1105
                           elif update_type == "rotation":
1106
1107
                                  self.weight_rotation = (
                                          self.updating_factor_weights * self.weight_rotation
1108
1109
                                          + (1 - self.updating_factor_weights) * self.lower_bound_weights
1110
                                  )
```

1111	def	<pre>optimize_LM(self, initial):</pre>
1112		unn
1113		Optimize the factor graph using Levenberg-Marquardt optimization of gtsam.
1114		Parameters
1115		
1116		initial : gtsam.Values
1117		Initial values of all variables.
1118		Returns
1119		
1120		result : gtsam.Values
1121		Uptimized values of all variables, including the estimated sensor pose.
1122		
1123		# aejine ine optimizer
1124		params = gtsam.Levenbergmarduardrarams()
1125		params.setverbosityLM("EKKUK")
1126		optimizer = gtsam.Levenbergmarquardtoptimizer(seif.graph, initiai, params)
1127		# optimize the graph
1128		result = optimizer.optimize()
1129		return result
1130	###	*****
1131	# H	elper functions
1132	###	***************************************
1133	def	get_construction_noise_with_degrees_of_freedom(
1134		<pre>self, first_symbol, second_symbol, axis_first_symbol</pre>
1135	):	
1136		unn
1137		Get the the construction noise for a part in a group with degree of freedom to other parts.
1138		Parameters
1139		
1140		first_symbol : gtsam.Symbol
1141		Currently investigated symbol.
1142		second_symbol : gtsam.Symbol
1143		Symbol of the part we want to investigate the connection to first_symbol with.
1144		axis_first_symbol : str
1145		Respective axis for the group with an degree of freedom.
1146		Returns
1147		
1148		construction_noise : str or float
1149		"dependent" if the two symbols are in the same group,
1150		otherwise the additional standard deviation for the noise is returned.
1151		
1152		
1154		second_symbols info[first_sumba]]["ground_with_degrees of freedow"][[svis_first_sumba]]
1154		. Sell.Symbols_inio[iiist_Symbol][ gloups_with_degrees_ol_ifeedom ][axis_iiist_Symbol] .
1156		/· return "dependent"
1157		else.
1159		return calf sumbals info[first sumbal]["std degraps of freedom"][avis first sumbal]
1158		Teturn Serr.Symbols_Info[IIISt_Symbol][ Stu_degrees_of_ITeedom ][axis_IIISt_Symbol]
1159	def	<pre>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):</pre>
1160		
1161		Get a noise model for the observations.
1162		Ine noise model is a diagonal matrix with the given standard deviations as entries.
1163		Parameters

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```
1164
1165
              roll : float, optional
1166
                  Standard deviation of the noise in the rotation around the x-axis. The default is 0.1.
1167
               pitch : float, optional
                  Standard deviation of the noise in the rotation around the y-axis. The default is 0.1.
1168
1169
               uaw : float. optional
                  Standard deviation of the noise in the rotation around the z-axis. The default is 0.1.
1170
1171
               x : float, optional
                   Standard deviation of the noise in the x-coordinate. The default is 0.3.
1172
1173
              y : float, optional
1174
                   Standard deviation of the noise in the y-coordinate. The default is 0.3.
1175
               z : float, optional
                   Standard deviation of the noise in the z-coordinate. The default is 0.3.
1176
               rot : float, optional
1177
1178
                  Standard deviation of the noise in the rotation around all axes.
1179
                  If rot is given, roll, pitch and yaw are set to rot. The default is None.
               pos : float, optional
1180
1181
                   Standard deviation of the noise in the translation in all directions.
1182
                   If pos is given, x, y and z are set to pos. The default is None.
1183
               Returns
1184
               noise_model : gtsam.noiseModel
1185
               Noise model for the observations.
1186
1187
1188
               if rot is not None:
                  roll, pitch, yaw = rot, rot, rot
1189
               if pos is not None:
1190
1191
                   x, y, z = pos, pos, pos
1192
               return gtsam.noiseModel.Diagonal.Sigmas(np.array([roll, pitch, yaw, x, y, z]))
1193
          def get_rotation_matrix(self, angle, axis="x"):
1194
1195
               Get the rotation matrix for a rotation around the x, y, or z axis.
1196
               Parameters
1197
               angle : float
1198
                 Angle of the rotation in radians.
1199
1200
               axis : str, optional
                  Axis of the rotation ("x", "y", or "z"). The default is "x".
1201
1202
               Returns
1203
               _____
               rotation_matrix : np.array
1204
1205
                  Rotation matrix.
1206
               if axis == "x" or axis == "roll":
1207
1208
                   return np.array(
1209
                       Г
                           [1. 0. 0].
1210
1211
                           [0, np.cos(angle), -np.sin(angle)],
1212
                           [0, np.sin(angle), np.cos(angle)],
                       ]
1213
                   )
1214
               elif axis == "y" or axis == "pitch":
1215
1216
                   return np.array(
1217
                      Ε
                           [np.cos(angle), 0, np.sin(angle)],
1218
1219
                           [0, 1, 0],
1220
                           [-np.sin(angle), 0, np.cos(angle)],
                       ٦
1221
```

```
1222
               )
            elif axis == "z" or axis == "yaw":
1223
1224
               return np.array(
1225
                  Ε
                      [np.cos(angle), -np.sin(angle), 0],
1226
                      [np.sin(angle), np.cos(angle), 0],
1227
1228
                      [0, 0, 1],
1229
                  ]
               )
1230
1231
            else: # incorrect input
1232
               print("Incorrect input for axis, please use x, y, or z!")
1233
               return None
1234
        def get_joint_rotation_matrix(self, angles, axes="xyz"):
1235
1236
            Get the rotation matrix for a rotation around the x, y, and/or z axis, angle in radians.
1237
            Parameters
1238
            angles : list of float
1239
              Angles of the rotation in radians around the axes specified in "axes".
1240
            axes : str. optional
1241
               Axes of the rotation (sequence of "x", "y", and "z"). The default is "xyz".
1242
1243
            Returns
1244
            _____
1245
            rotation_matrix : np.array
1246
              Rotation matrix.
1247
            rotation_matrix = np.eye(3)
1248
1249
            for i in range(len(angles)):
               rotation_matrix = np.dot(self.get_rotation_matrix(angles[i], axes[i]), rotation_matrix)
1250
1251
           return rotation_matrix
         .....
1252
        Additional functions, not shown here:
1253
            def get_dominant_part(self, list_of_parts)
1254
1255
            def get_width_loading_platform(self)
1256
            def get_current_sensor_pose(self)
         ......
1257
         1258
1259
         # Visualization
1260
         .....
1261
1262
         Visualization functions, not shown here:
1263
            def plot_sensor_trajectory(...)
            def plot truck(...)
1264
1265
            def create_cuboid(...)
1266
            def create_cylinder(...)
           def create_coordinate_frame(...)
1267
         .....
1268
     1269
     # Exemplary Test
1270
1271
     1272
     def test_model_error_construction_noise(
1273
         data_file, obs_noise_sd, constr_noise_sd, parameter_values, repeat_test
```

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):

1275	
1276	Test how fast and well the model of the truck is updated in dependence of the construction noise.
1277	Parameters
1278	
1279	data file : pickle file
1280	Pickle file containing the observed data in form of a dictionary.
1281	obs noise sd : float
1282	$S_{1}$ , $S_{2}$ , $S_{3}$ , $S$
1282	constrances of the float
1284	Standard deviation of the additional artificial noise for the construction of the truck
1285	normation welves - list of floot
1285	Different values of the construction noise used for testing
1280	Different dataes of the construction house used for testing.
1287	repeat_test : int
1288	States now often the test should be repeated.
1000	P. de sue
1289	Keturns
1290	
1291	means : naarray of shape (len(parameter_values), num_timesteps)
1292	Means of the model error for the different parameter values of all test trials.
1293	std_devs : ndarray of shape (len(parameter_values), num_timesteps)
1294	Standard deviations of the model error for the different parameter values of all test trials.
1295	"""
1296	# set a seed to get reproducible results
1297	np.random.seed(11)
1298	# get the observations from the data file
1299	<pre>observation_list = data_to_observations(data_file)</pre>
1300	# basic parameter settings
1301	observation_noise_translation_default, observation_noise_rotation_default = 0.17, 0.14
1302	construction_noise_rotation_default = 1.5
1303	updating_factor_noise, updating_factor_weights = 0.94, 0.97
1304	weight translation, weight rotation = 0.80, 0.80
1305	align every n steps = 32
1306	# create a basic instance of the class Truck3D with the parameters defined above
1307	Truck basic = Truck3D(
1308	observation noise translation default=observation noise translation default.
1309	observation noise rotation default=observation noise rotation default
1310	construction noise rotation default=construction noise rotation default
1311	undating factor noise undating factor noise
1319	undating factor weights=undating factor weights
1212	aparone_reverse translation_right translation
1313	initial_weights_translation-weight_translation,
1314	line outputs_forderon=weight_forderon,
1315	arrgn_every_n_steps-arrgn_every_n_steps,
1316	
1015	
1317	# list that will contain the model errors for all test trials
1318	results = []
1319	for _ in range(repeat_test):
1320	# copy the instance of the Truck3D class
1321	Truck_imprecise_configuration = copy.deepcopy(Truck_basic)
1322	# define an imprecise truck configuration with additional random noise
1323	<pre>constr_noise = np.random.normal(0, constr_noise_sd, 11)</pre>
1324	<pre>Truck_imprecise_configuration = get_imprecise_truck_configuration(</pre>
1325	Truck_imprecise_configuration, constr_noise
1326	
1327	# list that will contain the model errors for one test trial
1328	model_errors = []

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```
1329
               # add random noise to the observations
1330
              noisy_observations = add_noise_to_data(observation_list, obs_noise_sd)
1331
               # run the sensor pose estimations for all parameter values
              for j in range(len(parameter_values)):
1332
1333
                   # copy the imprecise Truck3D object
                   Truck = copy.deepcopy(Truck_imprecise_configuration)
1334
1335
                   # set the value of the construction noise
                   parameter = parameter_values[j]
1336
                   Truck.construction_noise_translation_default = parameter
1337
1338
                   # run the sensor pose estimation on these observations
                   model_error, _ = simulate_sensor_trajectory_model_error(Truck, noisy_observations)
1339
1340
                   model_errors.append(model_error)
1341
              results.append(model_errors)
          means = np.mean(results, axis=0)
1342
          std_devs = np.std(results, axis=0)
1343
1344
          return means, std_devs
1345
      def data_to_observations(data_file):
1346
           .....
          Get the observations saved in a data file.
1347
          Parameters
1348
1349
          _____
          data_file : pickle file
1350
              Opened pickle file containing the data.
1351
1352
          Returns
1353
          all_observations : list of dict
1354
              List containing the observations for each time step.
1355
          .....
1356
          data_dict = pickle.load(data_file)
1357
1358
          all_observations = []
          for frame in data_dict.values():
1359
              current_observation = {}
1360
1361
              for symbol in frame.keys():
                   pos, rot = frame[symbol]["pos"], frame[symbol]["rot"]
1362
1363
                   current_observation[symbol] = gtsam.Pose3(gtsam.Rot3(rot), pos)
1364
              all_observations.append(current_observation)
          return all_observations
1365
      def add_noise_to_data(data, noise_sd):
1366
           .....
1367
          Add random additional noise to data.
1368
          Parameters
1369
1370
1371
          data : list of dict
              Data of gtsam poses, where we want to add noise.
1372
1373
          noise_sd : float
```

1374		Standard deviation of the normally distributed noise.
1375		Returns
1376		
1377		noisy_data : list of dict
1378		Data of gtsam poses with random noise.
1379		nnn
1380		noisy_data = copy.deepcopy(data)
1381		for frame in noisy_data:
1382		for symbol_string in frame.keys():
1383		# get random numbers for the noise of position and orientation
1384		<pre>noise_pos = np.random.normal(0, noise_sd, 3)</pre>
1385		<pre>noise_rot = np.random.normal(0, min(noise_sd, 0.05), 3)</pre>
1000		
1386		# create a rotation matrix from the local coordinates noise_rot of SU(3)
1387		<pre>noise_rot_matrix = local_update_SU3(np.eye(3), noise_rot)</pre>
1200		# transform the noise to a steam mase
1280		$\pi$ trunsform the horize to a gradient pose
1389		noise_pose - gisam.roses(gisam.kots(noise_rot_matrix), noise_pos)
1390		# add the noise to the data
1391		<pre>frame[symbol_string] = frame[symbol_string].compose(noise_pose)</pre>
1392		return noisy_data
1393	def	local undate SD3(rotation matrix, vector):
1304		
1395		Local undate on $SO(3)$ using the hat-operator and the matrix exponential as a retraction
1000		
1396		Parameters
1397		
1398		rotation_matrix : ndarray of shape (3,3)
1399		Original rotation matrix RO.
1400		vector : ndarray of shape (3,)
1401		Vector xi of the local coordinates on SO(3).
		D. de una e
1402		Keturns
1403		
1404		naarray of snape (3,3)
1405		Local upaate of the original rotation matrix KU.
1406		notion notation matrix @ matrix and (matrix)
1407		return rotation_matrix @ matrix_exp(vector)
1408	def	<pre>matrix_exp(vector):</pre>
1409		"""
1410		Matrix exponential for a skew-symmetric matrix.
1411		Uses the hat-operator on SO(3) and computes the resulting matrix with Rodrigues' forumla.
a		Demonsterne
1412		rarameters
1413		
1414		vector : ndarray of shape (3,)
1415		Vector x1 of the local coordinates on SU(3).
1416		Returns
1417		
1418		exp(hat(vector)) : ndarray of shape (3,3)
1419		The matrix exponential of hat(vector).
1420		
1421		theta = np.linalg.norm(vector)
1422		if theta == 0:
1423		return np.eye(3)

```
1424
          else:
1425
               return (
1426
                   np.eye(3)
1427
                   + np.sin(theta) / theta * hat_operator(vector)
1428
                   + (1 - np.cos(theta)) / theta**2 * hat_operator(vector) @ hat_operator(vector)
               )
1429
      def hat_operator(vector):
1430
1431
           Implementation of the hat-operator for SO(3).
1432
1433
          Parameters
1434
           vector : ndarray of shape (3,)
1435
1436
              Vector xi of the local coordinates on SO(3).
1437
          Returns
1438
           hat(vector) : ndarray of shape (3,3)
1439
               Skew-symmetric matrix, output of the hat-operator.
1440
           .....
1441
1442
          return np.array(
1443
               Ε
                   [0, -vector[2], vector[1]],
1444
                   [vector[2], 0, -vector[0]],
1445
                   [-vector[1], vector[0], 0],
1446
               ]
1447
1448
          )
      def simulate_sensor_trajectory_model_error(Truck, observations):
1449
1450
1451
           Simulate a run of the sensor pose estimation for given observations.
          Parameters
1452
1453
            _____
           Truck : Truck3D
1454
              Instance of the Truck3D class with the respective parameters.
1455
           observations : list of dict
1456
              List of the noisy observations.
1457
1458
          Returns
1459
1460
           model_error : list of float
               List of the model errors for each time step.
1461
           .....
1462
1463
           # list containing the model errors for each time step
          model_error = []
1464
           # get the initial model error
1465
          model_error.append(evaluate_model_error(Truck))
1466
1467
          for i in range(len(observations)):
1468
               # estimate the sensor pose
               _, _, result = Truck.estimate_sensor_pose(observations[i])
1469
               # update the truck configuration according to the results of the previous estimation
1470
               Truck.update_truck_configuration(result)
1471
               # evaluate the current error of the estimated model
1472
1473
               model_error.append(evaluate_model_error(Truck))
```

```
1474
          return model error
```

```
1475
       def evaluate_model_error(Truck):
1476
            ......
```

```
Evaluate the model error of the truck. The model error is defined as the distance between
1477
           the poses of the parts in the model and the poses of the parts in the ground truth.
1478
           The distance in the position is computed as the norm of the difference between the positions.
1479
           The distance in the rotation R1, R2 is computed as the Frobenius norm of I\text{-}R1R2^{\text{-}}T.
1480
1481
          Parameters
1482
           Truck : Truck3D
1483
              Truck, where we look for the error between the assumed model and the ground truth.
1484
1485
          Returns
1486
1487
           cumulated\_distance : float
              Model error cumulated for all parts of the truck.
1488
           .....
1489
           cumulated_distance = 0
1490
           for symbol in Truck.symbols_info:
1491
               if symbol not in Truck.symbols_sensor:
1492
                   # compute the distance between the model and the ground truth
1493
                   pose_difference = Truck.symbols_info_to_pose_in_world_frame(
1494
1495
                       symbol, true_or_estimated="estimated"
1496
                   ).between(Truck.symbols_info_to_pose_in_world_frame(symbol, true_or_estimated="true"))
1497
                   cumulated_distance += pose_norm(pose_difference)
          return cumulated_distance
1498
1499
      def pose_norm(pose):
1500
1501
           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.
1502
1503
          Parameters
1504
           _____
          pose : gtsam.Pose3
1505
1506
```

```
Pose to compute the norm.
1507
           Returns
1508
1509
           norm : float
```

```
Norm of the given pose.
1510
           .....
1511
1512
          pose_norm_position = np.linalg.norm(pose.translation())
1513
          R = pose.rotation().matrix()
          I = np.identity(3)
1514
```

```
pose_norm_rotation = np.linalg.norm(I - R)
1515
```

```
1516
```

```
return pose_norm_position + pose_norm_rotation
       .....
1517
```

```
1518
      Additional functions, not shown here:
           def get_imprecise_truck_configuration(Truck, noise)
1519
1520
          visualization functions
1521
```

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