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3D robust stability polyhedron in multi-contact

Hervé Audren and Abderrahmane Kheddar

Abstract—We propose algorithms to compute the 3D robust stability region in multi-contact. It is well known that the stability region is a product of convex cones and hence is a convex polyhedron. Our stability region extends existing recursive 2D static stability approaches to 3D by accounting for possible center of mass accelerations. We provide algorithms that construct the region of robust stability in a systematic way. We compare our algorithms and discuss possible computation of intermediary shapes using morphing. Finally, we provide an example of usage in generating robust static postures that can serve the purpose of multi-contact planning.

Index Terms—Multi-contact, 3D Robust Static Stability.

I. INTRODUCTION

LEGGED robots have been substantially improved in the last years in terms of hardware and control. For legged robots, holding *stable* postures in multi-contact is critical to avoid falling under perturbations. Thus a criteria is necessary to enforce the stability of the equilibrium for a given contact stance/configuration. Ideally, this criteria would take the form of a convex function to be used in fast optimization programs. This function also depends on the bounded set of perturbations that is considered *a priori*: if the criteria is true, any perturbation within that set can be sustained. In that sense, our criteria will be *robust* w.r.t this set of perturbations.

The stability criteria are tightly linked to the dynamic equations that are known to be non-linear [19]. They establish a relation between joint torques, accelerations and contact forces through inertial parameters. Even when the whole-body dynamics is simplified to its center-of-mass (CoM) [41], [3], the angular momentum of the body is a cross-product between the contact forces and the CoM position; an operation that is neither linear nor convex.

Our idea is to compute a hull \mathcal{P} for the CoM noted c , such that $\forall c \in \mathcal{P}$, the stability is guaranteed to be *robust* w.r.t. a given convex set \mathcal{G} of CoM accelerations \ddot{c} , $\ddot{c} \in \mathcal{G}$. In fact, $\forall c \in \mathcal{P}$, there exist a set of contact forces that can generate any acceleration in \mathcal{G} . In other words, we compute

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the intersection of \mathcal{G} with the set of all possible motions. We show that this intersection results in a convex volume that we project in the CoM space. Hence, by defining \mathcal{G} , the CoM acceleration and its position are decoupled.

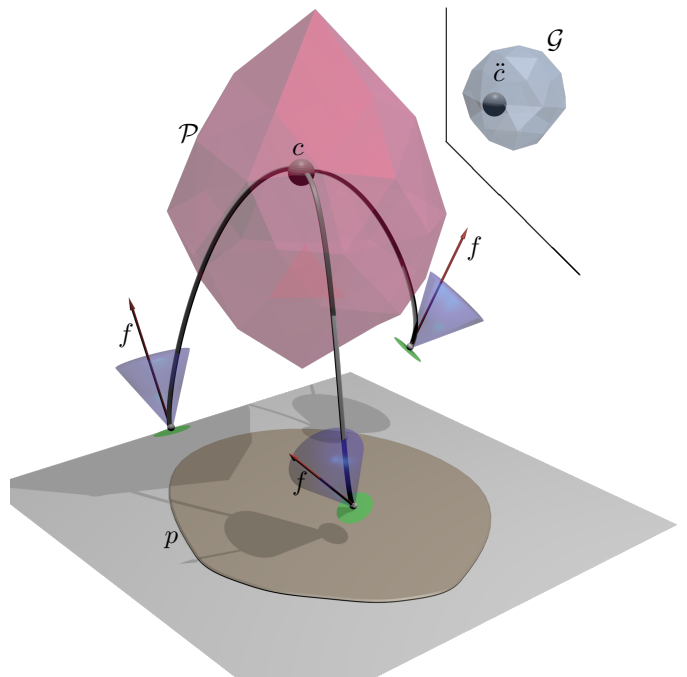


Fig. 1. Illustration of the problem annotated with the variables. We propose algorithms to compute the robust static stability \mathcal{P} for the CoM c , knowing the space \mathcal{G} of admissible CoM's accelerations \ddot{c} . The forces are all noted f , the set to which they belong. Contact surfaces are represented by the green disks with the contact points and their normals, represented by the blue friction cones. p is the static stability polygon as computed in [11].

In previous work, some restrictions have been set to obtain such a decoupling. One of the most stringent is to set the CoM acceleration to zero ($\ddot{c} = 0$), resulting in the static stability criterion [33]. Static stability (i.e. equilibrium) is used most notably in multi-contact posture generation [18], [9], [13].

Static stability can be a function of the gravity orientation. Indeed [32], [35] present a way to find all the gravity orientations (equivalently, all orientations of the assembly base) that satisfy static stability of assemblies: as it is a convex region defined by inequalities. They apply early vertex enumeration techniques—that we also use, to compute the acceptable region. Yet, the objects were fixed and not actuated.

The best known approach considers a virtual point that accounts for both the CoM position and its acceleration: the zero moment point (ZMP). The original ZMP criterion [46] is the extension of the *convex hull criterion* and has been widely used in biped locomotion on flat ground assuming high

friction, e.g. [29]. The ZMP is better defined in [23], and its similarities to the Center of Pressure (CoP) highlighted in [44]. The CoP is the local version of the ZMP criterion which allows it to be used in multi-contact but entails controlling independently the CoP at each contact area. Yet, the CoP is not defined when the normal force applied on the contact is nil, and is not easily extensible to bilateral contacts. Even with these drawbacks, CoPs was successfully applied in multi-contact control schemes such as in [45], [27], [47], [37] when combined with a CoM regulation policy for stabilization.

There is another multi-contact criteria: the resultant wrench of the contact forces must remain in a polyhedral convex cone. Pioneered by [43], it formed the basis of the work in [26] ostensibly titled “Adios ZMP”. It is applicable to multi-contact motion while remaining linear and global. This criterion is more clearly and properly established and applied to multi-contact motion in [15], [16]. In the latter works, a new ZMP-like criterion (a pseudo-ZMP) that has to remain in the two-dimensional projection of the convex polyhedral wrench cone is proposed. It is applicable to multi-contact motion and takes into account friction. Note that, similarly to the ZMP and the CoP, the pseudo-ZMP support area depends on the instantaneous CoM position. Instead, our approach imposes constraints on the resultant acceleration and finds a linear, global corresponding constraint on the CoM. It is an extension of our previous work in [2]: instead of computing regular static stability regions, we add explicit stability margins.

Computing wrench cones relies on the double description method [21] to perform projections. We observed that when the number of contacts increases slightly, the computation time increases drastically. Alternatively, *incremental projection* [30], computes an arbitrary close polyhedral approximation of convex projections without the need of having the full-dimensional polytope. As the static support area is a convex shape, this technique is fast and can be used in stability checks [11] and in multi-contact control [2]. Moreover, this technique does not require friction cones discretization. A very recent approach [40] suggests a quasi-analytic formulation of this region but it cannot handle arbitrary arrangements of contacts, and requires friction cones discretization in practice.

Constraining the CoM to remain in p is a global, CoM based, and linear criterion. Unfortunately, static stability (equilibrium) criterion does not imply dynamic stability [22], [2]. Indeed tracking a statically stable trajectory with changes in acceleration may induce falling. Thus, static stability is marginal, i.e. not robust to changes in the total acceleration.

We propose algorithms to compute a *robust* stability region \mathcal{P} for a given \mathcal{G} . Our approach has commonalities with [6], devised to check the robustness of a known trajectory and used linearized friction cones. [15] noted that such a region exists, but it was only sampled. In [39] a vertical cross-section of this region was computed using a line-sweep algorithm, not unlike that of [7] but of course not directly transposable to 3D.

Our approach is in a way the dual of the that shown in [14]: instead of limiting the CoM positions to a known convex polytope and finding the envelope of acceptable CoM accelerations, we propose to limit the CoM acceleration and find all acceptable CoM positions.

It is also more general than what is presented in [42]: the authors only consider errors on the contact forces without considering that the resulting acceleration could change.

Very recent works, developed in parallel to our approach have pointed at ways to compute the proposed region: [38] showed that when the CoM acceleration is not null, the CoM accessible region, nicknamed CFR, was a slanted prism, while [17] has shown how to use a polyhedral or ellipsoidal region, but this time to limit the CoM position.

We thus show the following:

- \mathcal{P} is a three dimensional convex shape (section III);
- It can be efficiently approximated by convex polyhedrons (section V). Also, if \mathcal{G} is limited to a convex polytope, it is the intersection of non-right prisms;
- Our construction algorithm can be modified to efficiently test the equilibrium of many points (section VII);
- Changes in acceleration can be well approximated by morphing polyhedrons (section VIII);
- We exemplify our stability criteria with a robust posture generation problem (section IX).

Now, we recall the recursive projection algorithm [11] that computes the static stability polygon p , represented in Fig. 1; this is because our work follows a similar methodology.

II. COMPUTATION OF THE STATIC STABILITY POLYGON

A. Recursive projection algorithm

The algorithm in [11] consists in building an inner and outer approximation of the true polygon, p , by solving a sequence of second-order cone programs:

$$\begin{aligned} \max_{c,f} \quad & d^T c \\ \text{s.t.} \quad & A_1 f + A_2(g)c = T(g) \\ & \|Bf\| \leq u(\mu)^T f \end{aligned} \quad (1)$$

where (see also Fig. 1 and appendix A),

- f is the set of contact forces;
- c is the CoM position;
- d is the given search direction;
- A_1 is the contact matrix that sums all forces and moments;
- A_2 represents the gravity wrench at position c ;
- B is the friction cone matrix;
- u is the matrix representing the contact normals and the friction coefficient μ ;
- T is the gravity wrench in function of the gravity g and the robot mass m ;
- $\|\cdot\|$ is the euclidian norm.

The solution c^* is an extremal CoM position in the direction d , and is thus added to the inner approximation. Conversely, the half-plane defined by $\{c \in \mathbb{R}^2 | d^T c > d^T c^*\}$ only contains infeasible points, and its complementary is a facet of the outer approximation. All points within the inner approximation are thus stable, those outside of the outer approximation are unstable, whereas the stability is undetermined for the points in-between the inner and outer approximations.

The difficulty of this class of algorithms is the choice of the search direction d . In statics, the undetermined region between

the inner and outer approximation is entirely composed of triangles. The next search direction is chosen to be perpendicular to the edge of the inner approximation that belongs to the triangle of maximum area and pointing outwards.

B. Problems associated with bilateral contacts

If we have opposite normals, there is no boundary on the forces applied along the pair of contacts line, and the problem is infeasible. Thus we limit the search region [2]:

- A bounding sphere on the CoM position: we add a cone constraint $\|c\| \leq c_{\max}$, which can be set from the kinematics limits of the robot;
- A limit on the forces: for each force, no single component should be greater (in absolute value) than the weight of the robot. As this is much of a heuristic, we rather set a limit on the actuator torque generated by the forces at each contact point.

For the sake of conciseness, we do not explicitly write those extra convex constraints in what follows. However, they are always present to act as safeguards in the implementations.

III. ROBUST STATIC STABILITY

The previously generated shape p gathers all the statically stable CoM positions for a given set of contacts. Now, we would like to know where the CoM can be when, in addition, we allow a given set of CoM accelerations¹, that keeps the forces within their friction cones. This is what we call the *robust stability region*, illustrated by \mathcal{P} in Fig. 1.

A. Problem formulation

Definition 1 (Robust static equilibrium). A CoM position c is in robust static equilibrium with respect to a given residual radius r iff:

$$\forall \tilde{g}_i \text{ such that } \|\tilde{g}_i\| < r, \exists f_i \text{ such that} \\ A_1 f_i + A_2(g + \tilde{g}_i)c = T(g + \tilde{g}_i) \quad (2)$$

$$\|B f_i\| \leq u^T f_i \quad (3)$$

All notations are the same as Equation 1. Recall that A_2 embeds the cross product with the gravity. In the non-robust case, its last line is nil as g is aligned with the vertical axis. Hence, the vertical component of c does not contribute anything, and eq. (1) defines a two-dimensional shape. In the robust case, $g + \tilde{g}_i$ is almost always *not* aligned with the vertical axis and eq. (2) defines a three dimensional shape. Moreover, this three-dimensional volume is an infinite intersection of convex shapes –one for each couple (f_i, g_i) , therefore it is a convex shape.

In [6], \mathcal{P} is a given point, of a predefined CoM trajectory, at which \mathcal{G} is computed. The latter is a sphere, of unknown radius r , centered on the acceleration that is obtained for a given contact forces. The authors present an algorithm to compute r based on gravito-inertial wrench projections and use it as a robustness measure for the given trajectory.

¹The said CoM accelerations can result from a perturbation, i.e. applied external forces, or from the control, i.e. from the actuators.

In [15] robust static stability was envisioned as the set of CoM positions \mathcal{P} , where a set of lateral accelerations \mathcal{G} , centered around g , could be generated by contact forces. Here, \mathcal{P} is only sampled, whereas our method systematically computes \mathcal{P} , which allows fast-testing of robust equilibrium of any points (see section VII).

Another approach in [42] proposes a set of algorithms, including a faster version of [11], to deal with postural robustness to contact force errors. That is to say, for a given uncertainty of the force, they seek if it can be balanced by the others. Yet, a point of the utmost importance was missed: in the robust case, the region describing all possible CoM positions is no longer a 2-dimensional polygon, but a 3-dimensional polyhedron as we described previously.

In Definition (1), we state the problem $\forall \tilde{g}_i$. In order to compute \mathcal{P} , it is necessary to reduce it to a tractable form. We do so using two different approaches. The first approach writes a more conservative constraint on eq. (2) and eq. (3). The second approach discretizes \mathcal{G} .

The goal of both approaches is to obtain a finite set of convex constraints that can be used in convex optimization problems. Combining this problem with the recursive projection algorithm [30], [11] allows to compute the corresponding convex shape. Furthermore, if those constraints are linear they can also be used in a direct projection algorithm as in [15].

B. Formulating a stricter constraint

In this section, we derive a new inequality (depending on r but not individual \tilde{g}_i), that will induce eq. (3). This new constraint has the same form as eq. (1). Let us consider the solution for the static problem:

$$A_1 f + A_2(g)c = T(g) \quad (4)$$

$$\|B f\| \leq u^T f \quad (5)$$

To do so, we introduce two bound vectors, l and s . Please refer to appendix B for details and proofs of the following statements.

Theorem 1. A suitable lower bound l is given by:

$$u^T f_i - u^T f \geq l = -\mu r m \bar{\sigma}(A_1^\dagger)(1 + \|c\|) \quad (6)$$

With \dagger the Moore-Penrose pseudo inverse and $\bar{\sigma}(A)$ the largest singular value of A .

Theorem 2. A suitable upper bound s is given by:

$$\|B f_i\| - \|B f\| \leq s = r m \bar{\sigma}(B A_1^\dagger)(1 + \|c\|) \quad (7)$$

Corollary 1. For a given CoM position c , if there exist contact forces f that satisfy:

$$A_1 f + A_2(g)c = T(g) \\ \|B f\| \leq u^T f - r m (\bar{\sigma}(B A_1^\dagger) + \mu \bar{\sigma}(A_1^\dagger))(1 + \|c\|) \quad (8)$$

Then c is in robust static equilibrium.

Proof. The proof is straightforwardly obtained by combining Theorem 2 and Theorem 1. \square

Unfortunately, eq. (8) is in general *not* tight, and the converse of Corollary 1 is not true. Indeed, let us examine under

which conditions we achieve tightness (refer to appendix B for more details):

- 1) \tilde{g}_i is collinear to the max singular vector of A_1
- 2) \tilde{g}_i is perpendicular to the max singular vector of A_1
- 3) \tilde{g}_i is perpendicular to c

The two first items are incompatible. Moreover, c will be different at each iteration, but A_1 is a constant which means that no single \tilde{g}_i can even saturate two of the above conditions.

Even if this constraint is the tightest conic constraint (in terms of L_2 -norm) we found, it is still conservative. However, it allows to reduce the dimensionality of the problem substantially as we only need to find a single set of forces associated to a CoM position.

C. Discretizing the hypersphere

Instead of the conservative bound of Definition 1 described previously, we approximate the sphere $\|\tilde{g}_i\| < r$ by a polytope whose k vertices are selected among \tilde{g}_i , that is:

$$g_i = g + \tilde{g}_i \quad i \in [0 \cdots k] \quad (9)$$

The closer this polytope is to a sphere of radius r , the closer the computed polyhedron will be to the real \mathcal{P} .

The projection of the acceleration on the horizontal directions is not null, thus angular momentum is generated along every axis. Hence, we still cannot use the method in [11]. Instead, we need to find 3D CoM positions that realize every k accelerations. That is, can we find k sets of forces $F = [f_0^T \cdots f_k^T]^T$ that produce $G = [g_0^T \cdots g_k^T]^T$ at a given CoM position c ?

Proposition 1. *If a CoM position c is in robust static equilibrium ($c \in \mathcal{P}$), there exists F that verifies:*

$$\begin{aligned} \Phi F + \Psi c &= \Gamma \\ \|\Lambda F\| &\leq \Upsilon F \end{aligned} \quad (10)$$

With:

$$\Phi = \text{diag}(A_1 \cdots A_1) \quad (11)$$

$$\Psi = [A_2(g_0)^T \cdots A_2(g_k)^T]^T \quad (12)$$

$$\Gamma = [T(g_0)^T \cdots T(g_k)^T]^T \quad (13)$$

$$\Lambda = \text{diag}(B \cdots B) \quad (14)$$

$$\Upsilon = [u \quad \cdots \quad u]^T \quad (15)$$

Proof. We stack k problems given by Definition 1, one for each g_i . As a robust CoM position verifies the constraints of Definition 1 for any \tilde{g}_i , it does also for any k of them. \square

With this approach, the robustness of the static equilibrium \mathcal{P} is obtained w.r.t. the polytope \mathcal{G} defined by the k \tilde{g}_i .

Remark. *Contrarily to the bounds-based approach, this does not rely on the fact that \mathcal{G} is spherical. Indeed, robustness is obtained w.r.t. any convex polytope \mathcal{G} , given by its vertices \tilde{g}_i .*

As we seek for an explicit representation of \mathcal{P} we need to project the presented systems of equalities and inequalities into the CoM space, \mathbb{R}^3 . In the next section, we show how the direct projection method is not suited to our particular problem. Efficient algorithms will be introduced in section V.

IV. DIRECT PROJECTION

\mathcal{P} is a 3D shape, and as such the 2D recursive projection algorithm presented in section II-A does not apply. A simpler method is direct projection, but it does not apply to our case. However, the tools and terminology used for direct projection are used in our algorithms presented in the following section.

To compute the projection of a certain set, the most straightforward idea is to compute the boundary of that set, then projecting this boundary yields the boundary of the projection. This is the fundamental principle of direct projection.

In our case, this means computing the boundary of the set defined by either eq. (8) or eq. (10). Such a problem is known as a *vertex enumeration problem*.

In the linear case, the existence of a solution is given by the Weyl-Minkowski theorem [48], [34]. It states that any convex region defined by a finite number of linear inequalities is a convex polyhedron with a finite number of vertices. Hence, any convex polyhedron can be equivalently represented in two ways:

- *hyperplanes representation* (H-rep): a matrix H composed of its normals and a vector b of associated offsets. A point a is interior to this region iff $Ha \leq b$;
- *vertices representation* (V-rep): a set of vertices V . A point a is interior to this region iff there exists a vector α having the size of V and with positive coefficients, such that $a = V^T \alpha$ with $\sum_i \alpha(i) = 1$.

A number of algorithms and implementations are available to perform the H-rep to V-rep mapping and *vice-versa*. They are referred to as the *double description* algorithms, and vary in their capabilities and performance. Yet, none of them is inherently superior to the others. Notable examples are Qhull [5] that computes convex hull and half-space representations from vertices using the Beneath-Beyond algorithm; the Cddlib [21] and the Parma Polyhedra Library (PPL) [4] both of which build on the Fourier-Motzkin elimination algorithm [20], [36], the dual of the Beneath-Beyond algorithm.

Using this tool, we lay out more precisely the procedure to compute the direct projection of a set defined by convex constraints. It is composed of four steps:

- 1) Reduce the problem to a finite set of linear inequalities and equalities;
- 2) Enumerate the vertices bounding that set (i.e. find the V-rep from the H-rep);
- 3) Project the V-rep in the given space, resulting in another set of vertices.
- 4) Compute the convex hull of the previously obtained set of vertices.

Alternatively, we can linearize eq. (10) by using a polyhedral approximation of the friction cones. Unfortunately, a linear \mathcal{G} with k vertices, leads to searching for k sets of forces. This is not an option since the double description has a worst case complexity that is exponential in the number of dimensions [25]. Plus, all iterative conversion algorithms are super-polynomial in the combined size of input and output, see [10], [28].

On the other hand eq. (8) is of low dimension, but has non-linear terms on both sides; its linearization is far from

being trivial. We thus investigate two alternatives: the *recursive projection* and the *prism intersection*.

V. COMPUTING THE ROBUST POLYHEDRON

A. Recursive projection

1) *Principle*: The recursive projection technique consists in approximating the projection of a convex set by linear boundaries [30]. We seek for the approximation of \mathcal{P} as a projection in the 3D space, of a higher dimensional space. The core process of the recursive projection is to iteratively select directions d . The furthest point found along d (the CoM c^*) will be part of the set (\mathcal{P}). The others, i.e. $\{c \in \mathbb{R}^3 | d^T c > c^*\}$ will be outside of the set (\mathcal{P}). The collection of c^* will form an inner (i.e. conservative) approximation of the set (\mathcal{P}), while the complementary of the collection of unacceptable points will be a convex outer (i.e. over) approximation. The boundary of the real set lies in-between those two approximations, and they match exactly whenever *all* directions have been enumerated. In practice, the number of iterations will be limited.

We also associate a stopping criterion that depends on a measure of the error between the inner and outer approximation. Also, for fast convergence, we have to choose suitable search directions.

We now present two ways to formulate an optimization problem that yields the appropriate c^* : (i) using bounds in subsection III-B and (ii) linearization in subsection III-C.

2) *Using bounds*: Unfortunately, the formulation (eq. (8)) obtained in subsection III-B is not suitable to an optimization problem. To turn it into a second-order cone program (SOCP), we need to transform eq. (8) into two single conic inequalities by introducing an additional slack variable ξ :

$$\|c\| \leq \xi \quad (16)$$

$$\|Bf\| \leq u^T f - rm(1 + \xi)(\bar{\sigma}(BA_1^\dagger) + \mu\bar{\sigma}(A_1^\dagger)) \quad (17)$$

Then, solving the following optimization problem gives us an extremal point:

$$\begin{aligned} \max_{c, f, \xi} \quad & d^T c \\ \text{s.t.} \quad & A_1 f + A_2(g)c = T(g) \\ & \|c\| \leq \xi \\ & \|Bf\| \leq u^T f - rm(1 + \xi)(\bar{\sigma}(BA_1^\dagger) + \mu\bar{\sigma}(A_1^\dagger)) \end{aligned} \quad (18)$$

3) *Using linearization*: It is much simpler as the constraints presented in subsection III-C are already in a form that can be combined with a linear cost function to form a SOCP:

$$\begin{aligned} \max_{c, F} \quad & d^T c \\ \text{s.t.} \quad & \Phi F + \Psi c = \Gamma \\ & \|\Delta F\| \leq \Upsilon F \end{aligned} \quad (19)$$

4) *The projection algorithm*: Given that we are dealing with convex 3-dimensional sets, a common measure of size is the volume. Thus we use as a stopping criterion the volume difference between the inner and outer approximations.

To maximize the error reduction at each step, we drew inspiration from [11] and choose the search direction to

be perpendicular to one of the facets of the current inner approximation. The facet we choose is that having the larger *volume* of the associated convex polyhedron ν , which is the set of points that are outside that facet but inside the outer approximation. This ν can also be seen as a cut of the outer approximation by this facet.

A given facet is defined by its normal n and its offset o —the distance to the origin, as $\{c \in \mathbb{R}^3 | n^T c \leq o\}$. Thus, the ν is easily computed if an H-rep of $\mathcal{P}_{\text{outer}}$, $(H_{\text{outer}}, b_{\text{outer}})$ is available:

$$\begin{aligned} \nu[\text{facet}] &= \{c \in \mathbb{R}^3 | c \in \mathcal{P}_{\text{outer}}, n^T c > o\} \\ &= \{c \in \mathbb{R}^3 | H_{\text{outer}} c \leq b_{\text{outer}}, n^T c > o\} \\ &= \text{CUT}(\mathcal{P}_{\text{outer}}, \text{facet}) \end{aligned} \quad (20)$$

In [11], the difference between the inner and outer approximation was a disjoint union of triangles. Therefore, computing their areas and adding and removing points and hyperplanes was easy: a simple list of ordered vertices is sufficient and all operations can be performed without a dedicated algorithm.

In our case, we need to use the double description to compute ν for each facet of the inner approximation. Indeed, one iteration starting from the simplest volume, a tetrahedron, does not generate a union of disjoint tetrahedrons, but an union of intersecting prismatic, see Figure 2.

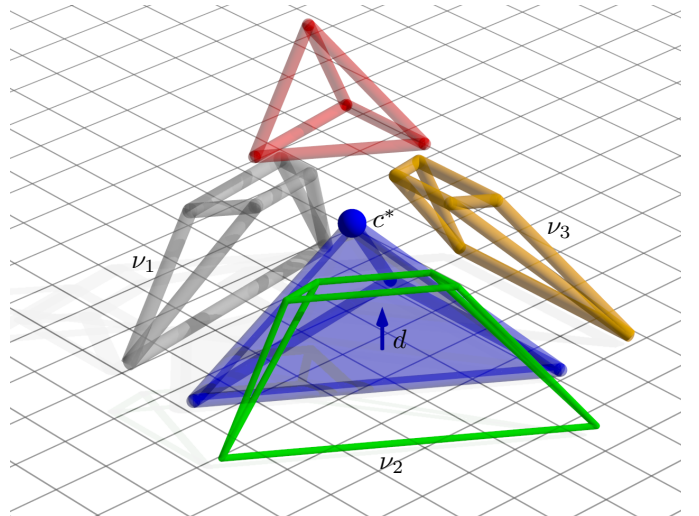


Fig. 2. Exploded view of one iteration of the algorithm for a simple example. The outer approximation is a tetrahedron. The search direction d is perpendicular to its base. The addition of the extremal point c^* forms the inner tetrahedron (transparent blue). The red pyramid is cut out of the outer approximation. The new uncertainty volumes ν associated to the new facets of $\mathcal{P}_{\text{inner}}$ are three prismatic with intersecting trapezoidal bases: thick orange ν_3 , thin green ν_2 , and transparent grey ν_1 .

For a given facet of the outer approximation, if we cut its ν with a parallel (to facet) hyperplane, there is no guarantee that this hyperplane does not intersect another uncertainty volume. Instead, we propose the following Algorithm 1:

- The input is the 6D contact positions and their friction coefficients.
- We initialize $\mathcal{P}_{\text{inner}}$, $\mathcal{P}_{\text{outer}}$ and the list ν of uncertainty volumes, by solving eq. (19) or eq. (18).

- We then select the direction d to be the normal to the facet having the biggest uncertainty volume.
- We solve the SOCP eq. (19) or eq. (18).
- The resulting c^* is added to the $\mathcal{P}_{\text{inner}}$, and the plane normal to this direction passing by c^* is added to $\mathcal{P}_{\text{outer}}$.
- For all facets, we compute their associated uncertainty volumes in two cases: (i) the facet was added at the last iteration and has no associated $\nu[\text{facet}]$ yet or (ii) its $\nu[\text{facet}]$ was intersected by the plane added at the previous iteration.

Algorithm 1 Robust polyhedron computation by recursive projection

```

1: contacts  $\leftarrow$  the set of contacts
2:  $\epsilon \leftarrow$  the approximation difference precision
3: procedure ROBUST(contacts,  $\epsilon$ )
4:    $\mathcal{P}_{\text{inner}}, \mathcal{P}_{\text{outer}}, \nu \leftarrow$  INIT(contacts)  $\triangleright$  (19) or (18)
5:   while  $\mathcal{V}(\mathcal{P}_{\text{outer}}) - \mathcal{V}(\mathcal{P}_{\text{inner}}) > \epsilon$  do
6:      $d \leftarrow$  normal(argmax(volumes))
7:      $c^* \leftarrow$  OPTIM( $d$ )  $\triangleright$  (19) or (18)
8:     plane  $\leftarrow$  PLANE( $d, c^*$ )
9:      $\mathcal{P}_{\text{inner}} \leftarrow \mathcal{P}_{\text{inner}} \cup c^*$ 
10:     $\mathcal{P}_{\text{outer}} \leftarrow \mathcal{P}_{\text{outer}} \cap$  plane
11:    for all facet  $\in \mathcal{P}_{\text{inner}}$  do
12:      if facet is new then
13:         $\nu[\text{facet}] \leftarrow$  CUT( $\mathcal{P}_{\text{outer}},$  facet)  $\triangleright$  (20)
14:      else
15:        if  $\nu[\text{facet}] \cap$  plane  $\neq \emptyset$  then
16:           $\nu[\text{facet}] \leftarrow$  CUT( $\nu[\text{facet}],$  plane)
17:        end if
18:      end if
19:      volumes[facet]  $\leftarrow \mathcal{V}(\nu[\text{facet}])$ 
20:    end for
21:  end while
22:  return  $\mathcal{P}_{\text{inner}}, \mathcal{P}_{\text{outer}}$ 
23: end procedure

```

A note on the polyhedral volume computation: computing the volume of a set of points is equivalently difficult to finding their convex hull. Indeed, to compute the volume of a random set of points, it is necessary to decompose it into a set of elementary volumes, the simplest being tetrahedrons. This is exactly what a convex hull of a set of points does. Similarly, finding the volume of a convex set defined by inequalities is equivalently difficult to enumerating its vertices. The convex hull operation is very well implemented in Qhull [5] when taking vertices as input. But, we found that its performance when taking inequalities as input is poor, so we used CDD [21]. However, Qhull was found to be slightly more numerically stable, and can be used in incremental mode, which makes it an attractive fallback. Still, numerical issues are inherent to the double description computations, and in pathological cases we switch to the PPL [4] that uses exact integer arithmetic and is much faster than CDD in this setting.

5) *Proof of convergence:* The following result holds.

Theorem 3. *The sequences $(\mathcal{P}_{\text{outer}})^{0, \dots, n}$ and $(\mathcal{P}_{\text{inner}})^{0, \dots, n}$ converge towards \mathcal{P} .*

Although it can seem obvious, we need to make sure that both the inner and outer approximations converge, that they have the same limit and that this limit is equal to \mathcal{P} . We thus introduce the following intermediate results.

Lemma 1. *$(\mathcal{P}_{\text{outer}})^n$ is monotonous non-increasing.*

Proof. $\mathcal{P}_{\text{outer}}^{n+1} = \text{CUT}(\mathcal{P}_{\text{outer}}^n, \text{plane})$ thus, $\forall a \in \mathcal{P}_{\text{outer}}^{n+1}, a \in \mathcal{P}_{\text{outer}}^n$ and $\mathcal{P}_{\text{outer}}^{n+1} \subseteq \mathcal{P}_{\text{outer}}^n$. \square

Lemma 2. *$(\mathcal{P}_{\text{inner}})^n$ is monotonous non-decreasing.*

Proof. $\mathcal{P}_{\text{inner}}^{n+1} = \text{CONV}(\mathcal{P}_{\text{inner}}^n, c^*)$ thus, $\forall a \in \mathcal{P}_{\text{inner}}^n, a \in \mathcal{P}_{\text{inner}}^{n+1}$ and $\mathcal{P}_{\text{inner}}^n \subseteq \mathcal{P}_{\text{inner}}^{n+1}$. \square

Lemma 3. $\forall n \in \mathbb{N}, \mathcal{P}_{\text{inner}}^n \subseteq \mathcal{P} \subseteq \mathcal{P}_{\text{outer}}^n$

Proof. $\forall n \in \mathbb{N}, c^* \in \mathcal{P}$. We denote $\pi_n = \{c | d_n^T c \leq c_n^*\}$. Thus, $\mathcal{P}_{\text{inner}}^n = \text{CONV}(c_i^* | i \in [0 \dots n]) \subseteq \mathcal{P}$. $\forall n \in \mathbb{N}, \pi_n^c \subseteq \mathcal{P}^c$. Thus $\mathcal{P}_{\text{outer}}^n = \bigcap_{0 \dots n} \pi_n \supseteq \mathcal{P}$. \square

We can now prove our main result:

Proof of Theorem 3. By the Lemmas 1 to 3, the sequences are monotonous and bounded: they converge. Thus, we can consider their limits, $\mathcal{P}_{\text{outer}}^\infty$ and $\mathcal{P}_{\text{inner}}^\infty$. At this limit, we have:

- $c^* \in \mathcal{P}_{\text{inner}}^\infty$
- $\pi \cap \mathcal{P}_{\text{outer}}^\infty = \mathcal{P}_{\text{outer}}^\infty$

Thus, in that direction d , $\mathcal{V}(\nu_d) = 0$. As we choose the maximum volume for the stepping direction, $\max_{\text{facet}} \mathcal{V}(\nu) = 0$. Then, $\forall \text{facet}, \mathcal{V}(\nu_k) = 0$. As the difference between $\mathcal{P}_{\text{outer}}$ and $\mathcal{P}_{\text{inner}}$ is exactly $\mathcal{V}(\bigcup \nu_k)$, $\mathcal{V}(\mathcal{P}_{\text{outer}}^\infty \setminus \mathcal{P}_{\text{inner}}^\infty) = 0$. As both $\mathcal{P}_{\text{inner}}^\infty$ and $\mathcal{P}_{\text{outer}}^\infty$ are non-empty convex sets $\overline{\mathcal{P}_{\text{inner}}^\infty} = \overline{\mathcal{P}_{\text{outer}}^\infty}$.

$\mathcal{P}_{\text{outer}}^\infty = \left(\bigcup_{i \in \mathbb{N}} \pi_n^c \right)^c$, thus $\mathcal{P}_{\text{outer}}^\infty$ is closed, and we have $\overline{\mathcal{P}_{\text{inner}}^\infty} = \mathcal{P}_{\text{outer}}^\infty$. As the sequence of c^* is included in \mathcal{P} , it is bounded, and thus compact. Then, as $\mathcal{P}_{\text{inner}}^\infty = \text{CONV}(c_n^* | n \in \mathbb{N})$, $\mathcal{P}_{\text{inner}}^\infty$ is also compact, and we get $\overline{\mathcal{P}_{\text{inner}}^\infty} = \mathcal{P}_{\text{inner}}^\infty = \mathcal{P}_{\text{outer}}^\infty$, which entails $\mathcal{P}_{\text{inner}}^\infty = \mathcal{P}_{\text{outer}}^\infty = \mathcal{P}$ by Lemma 3. \square

B. An intersection of prisms

The 3-dimensional shape described in Equation 18 is the intersection of k shapes. We show that each of these shapes is a prism, whose base can be computed as in subsection II-A.

Let us consider \tilde{g}_i and the 3D vector, c_0 that describes the 2D CoM position in the plane $c_z = 0$:

$$c_0 = \begin{bmatrix} 1 & 0 & 0 \\ 0 & 1 & 0 \\ 0 & 0 & 0 \end{bmatrix} c \quad (21)$$

Let's find which shape is described by the constraints:

$$\begin{aligned} A_1 f + A_2(g + \tilde{g}_i)c &= T(g + \tilde{g}_i) \\ \|Bf\| &\leq u^T f \end{aligned} \quad (22)$$

Note that A_2 is a function of $g_i = g + \tilde{g}_i$, and it is the only part of the equation that affects c . We know that:

$$A_2 c = \begin{bmatrix} 0 \\ -T(mg)c \end{bmatrix} = -m \begin{bmatrix} 0 \\ g \times c \end{bmatrix} \quad (23)$$

Thus, for any $c = [c_x \ c_y \ c_z]^T$:

$$T(mg)c = m \begin{bmatrix} -g_i^z c_y + g_i^y c_z \\ g_i^z c_x - g_i^x c_z \\ -g_i^y c_x + g_i^x c_y \end{bmatrix} \quad (24)$$

$$= m \begin{bmatrix} -g_i^z c_y \\ g_i^z c_x \\ -g_i^y c_x + g_i^x c_y \end{bmatrix} + m \begin{bmatrix} g_i^y c_z \\ -g_i^x c_z \\ 0 \end{bmatrix} \quad (25)$$

$$= T(mg)c_0 + m \begin{bmatrix} g_i^y c_z \\ -g_i^x c_z \\ 0 \end{bmatrix} \quad (26)$$

Thus we can show that the solution at any altitude is the translated of the solution at $c_z = 0$. Indeed by using eq. (26),

$$T(mg) \left(c_0 - \begin{bmatrix} \frac{g_i^x}{g_i^z} c_z \\ \frac{g_i^y}{g_i^z} c_z \\ 0 \end{bmatrix} \right) = T(mg)c \quad (27)$$

Thus, for any \tilde{g}_i , the shape described by Equation 22 is an infinite prism, whose base can be computed by finding the 2D polyhedron described with a slightly different problem:

$$\begin{aligned} \max_{c_2, f} \quad & d^T c_2 \\ \text{s.t.} \quad & A_1 f + A_2 \phi c_2 = T(g + \tilde{g}_i) \\ & \|Bf\| \leq u^T f \end{aligned} \quad (28)$$

where c_2 is the 2-dimensional CoM position in the plane $c_z = 0$ and ϕ is a projection matrix:

$$\phi = \begin{bmatrix} 1 & 0 & 0 \\ 0 & 1 & 0 \end{bmatrix}^T \quad (29)$$

The axis of this prism is thus given by the ρ_i , collinear to the g_i :

$$\rho_i = \begin{bmatrix} \frac{g_i^x}{g_i^z} & \frac{g_i^y}{g_i^z} & 1 \end{bmatrix}^T = \frac{1}{g_i^z} g_i \quad (30)$$

Then, the resulting intersection of prisms can be easily computed as follows:

- 1) Compute an approximation of each base using eq. (28) and the recursive projection in subsection II-A.
- 2) Compute the convex hull of each prism using a fixed height (possibly large) for the prism.
- 3) Compute the intersection of the prisms. It is the V-rep of the stacked H-rep of each prism.

The resulting intersection is illustrated in fig. 3.

VI. COMPARATIVE RESULTS

We tested the computation on a simple 3-contact scenario, using an acceleration polytope \mathcal{G} with 4 vertices. It is a “lozenge” whose vertices are generated from adding and removing an acceleration g_m to each lateral component of the gravity, that is $g_i = [\pm g_m, \pm g_m, g]$. We use Algorithm 1 with the linearization (subsection III-C): at each step, the problem contains 39 variables; 36 linear constraints that limit the maximum amplitude of each force component; 13 conic

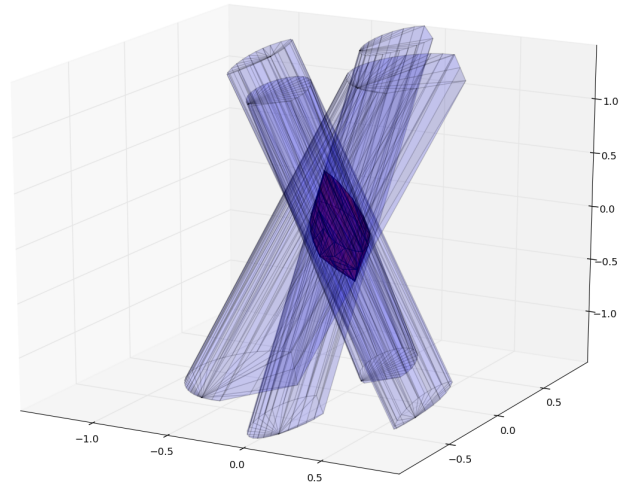


Fig. 3. The robust static stability polyhedron as an intersection of oblique prisms, for a residual radius $r = 3.55 \text{ m s}^{-2}$

constraints that enforce friction and limit the CoM to a sphere of unit radius.

The resulting \mathcal{P} s are presented in Figure 4. The obtained \mathcal{P} s are not right prisms as those of the non-robust case (displayed for reference in Figure 6): they have a diamond-like shape.

When using the acceleration sphere developed in subsection III-B, the polyhedron shrinks faster, see Figure 5. Indeed, the constraint eq. (8) is isotropic: only the norm of c intervenes. Hence, as the residual radius increases, the polyhedron shrinks in all directions at the same rate. Because eq. (8) uses maximum singular values, this shrinking rate is high. This shows that eq. (8) is far from being tight, and should only be used when fast computation is paramount.

Indeed Algorithm 1 converges in around 1 s to an acceptable precision of 0.043 m^3 (1.02%), see Figure 7. The performance we obtained is far from optimal: we used² Python and the free solver CVXOPT but it is known that switching to a compiled language such as C++ and using commercial solvers could greatly improve the runtime. The computation time can be split in two: (i) solving the optimization problems *per se*, and (ii) other operations.

The problem eq. (18) has a constant number of variables, whereas the problem eq. (19)’s size depends on the number of vertices of \mathcal{G} . As the other parts of Algorithm 1 do not depend on the method of discretization, we compare the computation times of the optimization problems in Table I. It shows that reducing the number of variables makes a great difference in favor of eq. (18). This also proves that it is very important to limit the number of the double-description operations.

What is not apparent in Figure 7 is that the computation time increases with the number of iterations. Indeed, as the algorithm progresses, more volumes have to be compared and potentially recomputed to find the best direction.

To verify that our construction is correct, we compare those polyhedrons to the naive sampling approach. On the one hand, we compute the polyhedrons for 20 residual radiuses

²<https://github.com/haudren/stabilipy>

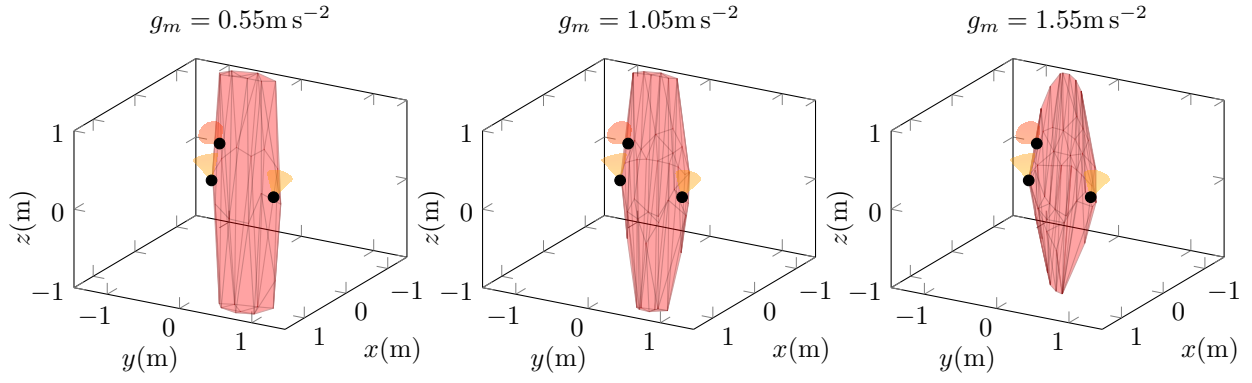


Fig. 4. Comparison of different \mathcal{P} for different acceleration polytopes \mathcal{G} generated by g_m . Only the inner approximations $\mathcal{P}_{\text{inner}}$ are rendered (in red), but are almost superimposed with $\mathcal{P}_{\text{outer}}$. The black dots with yellow cones represent contact points with associated friction cones.

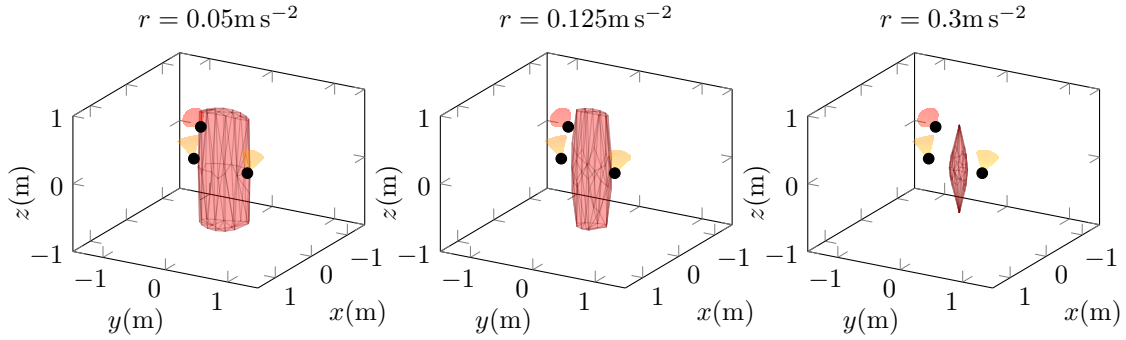


Fig. 5. Comparison of different \mathcal{P} for different acceleration spheres of radius r . Only the inner approximations $\mathcal{P}_{\text{inner}}$ are rendered (in red), but are almost superimposed with $\mathcal{P}_{\text{outer}}$. The black dots with yellow cones represent contact points with associated friction cones.

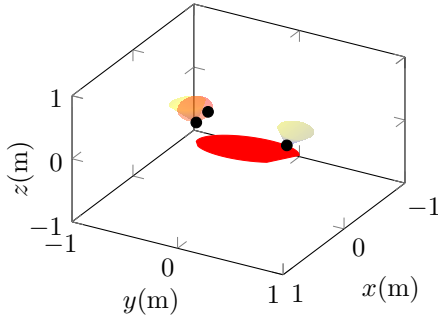


Fig. 6. Stability polygon in 2D (in red). The black dots with yellow cones represent contact points with associated friction cones.

linearly spaced in the interval $[0.55, 3.55]$ and for different linearizations of a spherical \mathcal{G} , from 4 to 18 vertices. On the other hand, we randomly select 100000 CoM positions c_t in the bounding box of each polyhedron, and test them for robust stability by solving eq. (19) at constant $c = c_t$ for an 18-vertices approximation of a spherical \mathcal{G} .

To compare those two objects, we introduce the symmetric difference metric:

$$\delta(\mathcal{P}_1, \mathcal{P}_2) = \frac{\mathcal{V}(\mathcal{P}_1) + \mathcal{V}(\mathcal{P}_2) - 2\mathcal{V}(\mathcal{P}_1 \cap \mathcal{P}_2)}{\mathcal{V}(\mathcal{P}_1) + \mathcal{V}(\mathcal{P}_2)} \quad (31)$$

This metric is the volume of the symmetric difference

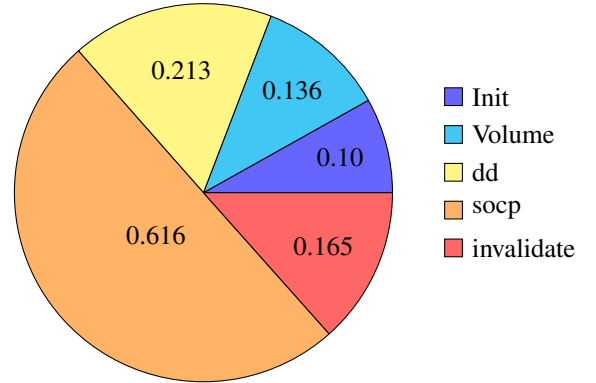


Fig. 7. Computation times for 3 contacts, polytope with 4 vertices and 3 point contacts. 50 iterations, total time 1.29s (clock time, without interpreter set-up and teardown). Final error: 0.043 m^3 .

$\mathcal{P}_1 \ominus \mathcal{P}_2$ normalized by the sum of volumes. Thus, we compare the convex hull of the stable c_t with the computed polyhedrons using the symmetric difference metric in Figure 8. It shows that at high number of vertices, and at any residual radius, the difference between the two objects is very small. This relative difference increases as the residual radius increases: as the polyhedron gets smaller, the bounding box shrinks, and thus the sampling density increases. Thus the sampling becomes more precise but not our approximation. Moreover, as we

TABLE I
COMPUTATION TIME OF 50 SOCP PROBLEMS WHEN CONSIDERING A SPHERICAL \mathcal{G} COMPARED TO THE LINEARIZATION. IN THE LATTER CASE, IT IS A FUNCTION OF THE NUMBER OF VERTICES OF \mathcal{G} .

Number of vertices	Time (s)
Spherical (1)	0.372
4	0.628
6	1.23
8	1.70
10	2.58
14	5.01
18	8.22

normalize by smaller quantities, this effect is amplified. Still, some error remains: because we only compute \mathcal{P} to a certain accuracy, some points that fall in the undetermined region are outside our approximation \mathcal{P} but still stable. This is also the reason why using 10 vertices seems slightly better than 18: the 10-vertex polytope corresponds to a slightly bigger \mathcal{P} that compensates for the width of the undetermined region.

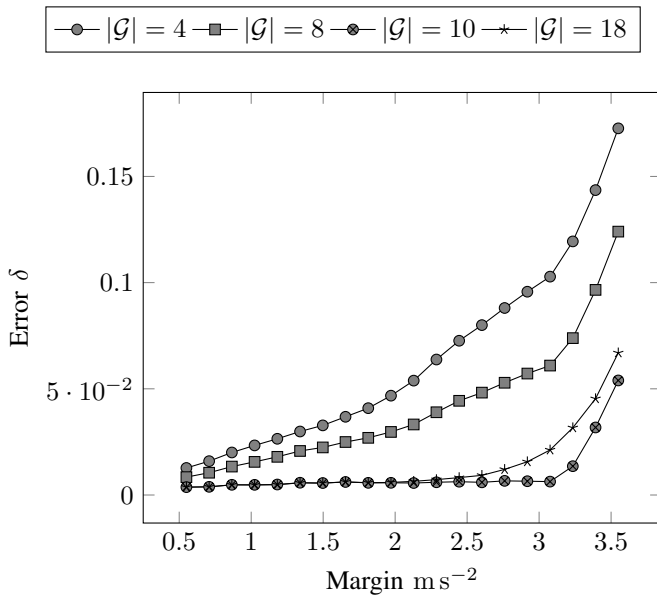


Fig. 8. Comparison between the convex hull of sampled points and the computed \mathcal{P} at various residual radiuses for different number of vertices of \mathcal{G} . Sampled points were tested for stability with $|\mathcal{G}| = 18$

To assess the convergence characteristics of our algorithm, we devised a worst-case scenario: \mathcal{P} would be a sphere (we took 6 contacts, one per each axis plus a limitation on the CoM or the contact forces). The results in Figure 9 show that the convergence is almost linear in the number of iterations. This does not compare to the nice quadratic result of [11], which is expected because our problem is 3D and not 2D. Indeed, it is proven in [24], [8] that the error δ between a smooth N -dimensional convex body \mathcal{P} and its best n -vertices polyhedral approximation asymptotically ($n \rightarrow \infty$) behaves as:

$$\delta \sim \gamma(N, \mathcal{P}) \frac{1}{n^{\frac{2}{N-1}}} \quad (32)$$

with γ a constant that only depends on the dimension N and the shape of the approximated volume \mathcal{P} .

As we are computing an approximation of the sphere, we cannot expect to converge faster than the best approximation. This point is further discussed in appendix C.

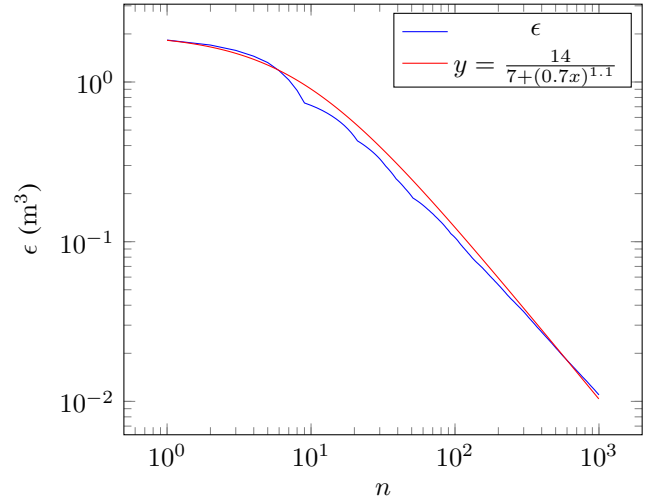


Fig. 9. Precision reached as a function of the number of iterations while approximating a sphere.

It is faster to compute \mathcal{P} using the prism intersection method (subsection V-B). First, at each iteration we solve k SOCP, each of which is approximately $\mathcal{O}(n^3)$ in the number of variables; whereas in Algorithm 1, we solve at each iteration a single problem whose size grows with k and has a complexity of about $\mathcal{O}((nk)^3)$. Second, we need $\mathcal{O}(\frac{1}{\epsilon})$ iterations to reach the desired precision ϵ in Algorithm 1, while only $\mathcal{O}(\frac{1}{\sqrt{\epsilon}})$ is needed using the prism intersection method. Last, with the prism intersection method, we do not need to keep track of the uncertainty volumes ν . In practice, it takes about 0.83 s to compute the same example as Figure 7 with a similar precision using the prism intersection method; in this case, most of the time is spent solving optimization problems. The intersection computation is not significant. Another advantage in using prisms is its robustness to numerical errors. On the other hand, it is not possible to target exactly a desired precision ϵ . Indeed, we do not know how large the intersection of prisms will be before computing it.

VII. INCREMENTAL PROJECTION FOR TESTING

A. Why incremental projection?

Testing if a particular point c_t is statically stable, can be done by solving the SOCP derived from Equation 1 as follows:

$$\begin{aligned} \max_f \quad & 1 \\ \text{s.t.} \quad & A_1 f = T(g) - A_2 c_t \\ & \|Bf\| \leq u^T f \end{aligned} \quad (33)$$

Hence, we would need to solve as many SOCP as there are CoM to be tested. To circumvent this, one can leverage the fact that a CoM position is in robust static equilibrium iff it is

inside the robust static polyhedron. To do so, a TEST-SAMPLE routine was introduced in [11] for the strictly static case. Their idea is to test if points are inside/outside of the static stability polygon, or fall in-between the inner and outer approximation. In that latter case, it is necessary to refine the approximations until the position of c_t is determined.

The difference between TEST-SAMPLE and the polygon computation in subsection II-A is only in the choice of the search direction: the point being tested is outside one (and only one) of the facets of the inner approximation. Hence, the search direction is chosen perpendicular to that facet.

By doing so, all the points are tested by two matrix multiplications. The result of the latter operation can tell us if points are in or out the stability polygon, otherwise they reside in-between and need refinement. This leads to an interesting observation: we can start with a very rough approximation of the stability polygon and refine it with the tests as needed.

We present how to adapt this methodology to $c_t \in \mathcal{P}$.

B. Polyhedral case

We need to update Algorithm 1 for testing. In 3D, testing $c_t \in \mathcal{P}$ can also be done by two matrix multiplications. However, when we need to update the approximations, c_t may lie outside of *multiple* facets of the $\mathcal{P}_{\text{inner}}$. Randomly selecting a facet gives us a direction. Hence, we get rid of the bookkeepings related to maintaining a map of the facets to their corresponding uncertainty volumes ν .

On the one hand, the direction can be selected to be perpendicular to the facet of $\mathcal{P}_{\text{inner}}$ forming the greatest uncertainty volume ν containing c_t . This however does not present any significant reduction in the number of iterations (i.e. the number of optimization problems being solved), but adds overheads due to the volume computations.

We thus choose a random selection for its efficiency.

C. Prism case

To test a point while using the prisms intersection method in subsection V-B, we apply Algorithm 2, derived from Algorithm 1. It takes as input a set of polygons, $\{p_i\}$ that contain an inner and outer approximation. Each of these polygons uses a different \tilde{g}_i and is computed using Equation 28.

For every operation, we do *not* compute the intersection of prisms; we only store the polygons that form their bases.

To test if a point is inside or outside one of the prisms, we compute its oblic projection along the prism axis, ρ_i , on the plane $c_z = 0$. We denote ψ the projection operator:

$$c_2 = \phi(c_t - \rho_i c_t^z) = \phi c_t - \frac{c_t^z}{g + \tilde{g}_i^z} \begin{bmatrix} \tilde{g}_i^x \\ \tilde{g}_i^y \\ \tilde{g}_i^z \end{bmatrix} \triangleq \psi(\tilde{g}_i) c_t \quad (34)$$

ϕ is defined in Equation 29.

Then we check if c_2 is inside every polygon, or outside any polygon. If it is neither inside nor outside, we refine the appropriate polygons until c_2 is determined. If it cannot, within a predefined iterations maxIter , we reject it. In fact maxIter is reached only when a point is exactly on the boundary. Alternatively, the area of the current triangle can be used as a stopping criteria whenever it becomes very small.

Algorithm 2 Algorithm to test a sample with the prism case while iteratively improving the underlying polygons

```

1: procedure TEST-PRISMS( $\{p_i\}, \{\tilde{g}_i\}, c_t, \text{maxIter}$ )
2:   nrIter  $\leftarrow$  0
3:   while nrIter  $\leq$  maxIter do
4:     if  $c_t \in \cap \{p_i^{\text{inner}}\}$  then
5:       return true,  $\{p_i\}$ 
6:     else if  $c_t \notin \cap \{p_i^{\text{outer}}\}$  then
7:       return false,  $\{p_i\}$ 
8:     else
9:       for  $p_i, \tilde{g}_i \in (\{p_i\}, \{\tilde{g}_i\})$  do
10:         $c_2 \leftarrow \psi(\tilde{g}_i) c_t$ 
11:        if  $c_2 \in p_i^{\text{outer}}$  and  $c_2 \notin p_i^{\text{inner}}$  then
12:           $p_i \leftarrow \text{TEST-SAMPLE}(p_i, c_2)$ 
13:        end if
14:      end for
15:    end if
16:    nrIter ++
17:  end while
18:  return false,  $\{p_i\}$ 
19: end procedure

```

D. Results

We use the examples in section VI for testing one million uniformly sampled random points in a box of dimensions $1 \text{ m} \times 1 \text{ m} \times 3 \text{ m}$. We used the incremental version of Algorithm 1 presented in subsection VII-B. The results are illustrated in Figure 10. Note that as the number of query points increases, the number of the iterations –to determine whether they are robustly stable or not, increases but at a sub-linear rate. The horizontal lines represent the number of iterations necessary to reach a certain precision ϵ using Algorithm 1.

For testing, it is faster to use the polyhedral version rather than the prism intersection one. Indeed, when testing many samples, the performance bottleneck is no longer the number of iterations, but rather the time necessary to test a sample. In the polyhedral case, two matrix multiplications at most are necessary to test a point, while in the prism intersection, at most $2k$ multiplications are necessary. Moreover, to test a point in the prism intersection, it is necessary to refine all polygons in which the projected point lands between the inner and outer approximation until it is outside of at least one of them or all of them contain it.

On a mono-core, Python implementation, it takes about 40s to test one million points, but this could of course be greatly improved using parallelization. Parallel computing is impractical to build \mathcal{P} in Algorithm 1 because each iteration will modify the approximations, but in the testing case, most samples can be tested independently.

VIII. EXTENSIONS AND DISCUSSIONS

A. Choice of the method

So far we have two methods to compute a robust static polyhedron \mathcal{P} : (i) recursive projection and (ii) prism intersection.

Whenever a low-precision approximation is sufficient, both methods are nearly equivalent. For high-precision (at least

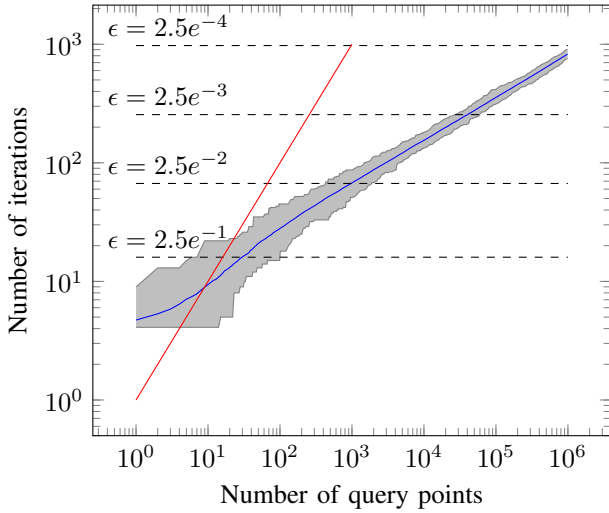


Fig. 10. Number of iterations required to test random points compared to naive sampling. The blue line represents the average number of iterations, the shaded area corresponds to the minimum-maximum envelope. The red line represents the naive sampling: one iteration per query. The dashed lines represent the number of iterations necessary to reach the indicated precision using algorithm 1.

30 iterations), we advocate prism intersection as it converges faster and has better numerical behaviour.

For testing, we similarly think that for few points, both methods are applicable. However, when testing a large number of points (at least 30), we strongly suggest using recursive projection for its speed advantage.

B. Morphing and change in robustness

We know how to compute \mathcal{P} corresponding to a given \mathcal{G} , which allows us to know which CoM positions are robust w.r.t \mathcal{G} . However, we have no means of knowing if those CoM positions are robust w.r.t. another polytope. We will explore how polyhedron morphing can approximate changes in \mathcal{G} .

More precisely, having $(\mathcal{G}_0, \mathcal{P}_0)$ and $(\mathcal{G}_1, \mathcal{P}_1)$, can we compute \mathcal{P}_λ from $\mathcal{G}_\lambda = \lambda_0 \mathcal{G}_0 + (1 - \lambda) \mathcal{G}_1$, $\lambda \in [0, 1]$?

Both $\mathcal{P}_0, \mathcal{P}_1$ are the projection of a high-dimensional convex shapes that lives in the manifold of the contact forces and the CoM position. Thus, deforming \mathcal{G} induces non-linear changes that are projected in the 3-dimensional space.

We propose to use linear morphing between convex polyhedrons to approximate changes in the acceleration polytope \mathcal{G}_λ . Hence, we consider scaling of the acceleration polyhedron, i.e. changes in the residual radius r but not in the linearization. To compute \mathcal{P}_λ , we use the morphing algorithm presented in [31]. Although the latter presents strategies to morph non-convex polyhedrons, at its core the algorithm is a 5-step process to find a common topology of two convex polyhedrons:

- 1) Get two polyhedrons (vertices and associated topology) and translate them to barycentric coordinates. Project them on the unit sphere.
- 2) Using neighbouring information to limit the search space, find every intersection of all spherical edges of the projections.

- 3) Order the intersections according to topological information and find in which facet of one polyhedron lie the vertices of the other.
- 4) For each vertex on the unit sphere, use barycentric coordinates to find its position on the original polyhedron.
- 5) Output the combined topology.

Then, we obtain intermediate polyhedrons by linearly interpolating between matching topologies. The Figure 11 shows an example of the results obtained by this algorithm.

We compare the volumes obtained in Figure 4 with those resulting from the morphing. To compare the volumes, we select an interval $[r_l, r_u]$ and a number of percentages λ . For each point we compute the exact polyhedron \mathcal{P}_e corresponding to the discretization of the robust sphere of radius $r = (1 - \lambda)r_l + \lambda r_u$ and the interpolated polyhedron at λ , \mathcal{P}_λ .

Figure 12 shows how $\delta(\mathcal{P}_e, \mathcal{P}_\lambda)$ evolves as a function of r , given different interpolation keypoints. The key observation is that morphing is a good fit (less than 12%) error as long as the distance $r_u - r_l$ between keypoints is less than 2 m s^{-2} . The problem becomes infeasible above $r = 4.0 \text{ m s}^{-2}$ meaning that computing 3 keypoints is enough to cover the whole acceptable range of residual radiuses. However, the interpolated polyhedron is *bigger* than the actual one, which may lead to false positives when testing for robust stability.

C. Robust 2D and robust 3D

We present a serious limitation of the 2-dimensional robust approaches [39], [42]: the robust static region is a 3-dimensional polyhedron, but they consider a 2-dimensional region, and then extend it to a right prism. We show that this approximation is a poor fit of the real volume, that only gets worse as the residual radius increases.

In order to compare our method to 2-dimensional approaches, we compare the volumetric computation to right (vertical) prisms whose bases are polygons. The right prism will have the same height as the diameter of the limiting sphere we use to bound the CoM accessible region, c_{\max} .

We first ensure that the volume that we compute is always included in the prism formed by the non-robust static stability polygon. We then compute a “robust static stability polygon” at height h , i.e. the recursive projection defined by:

$$\Phi F + \Psi \left(\phi c_2 + [0 \ 0 \ h]^T \right) = \Gamma \quad (35)$$

$$\|\Delta F\| \leq \Upsilon F \quad (36)$$

where c_2 is a 2D vector, ϕ is defined in Equation 29.

We use our previous examples, i.e. the same contacts and \mathcal{G} than in section VI, also for 20 residual radiuses linearly spaced in the interval $[0.55, 3.55]$, and for five different heights we compute both forms (right prism and \mathcal{P}). We then compute the error metric δ between the right prism and the 3D volume at each residual radius. We ensure that both computations lead to the same result by comparing the intersection of \mathcal{P} and the plane $c_z = h$ with the robust static stability polygon (2D). We use the same error metric in eq. (31), replacing the volume with the area. An example of the objects being compared is in Figure 13 while results are shown in Figure 14. The area

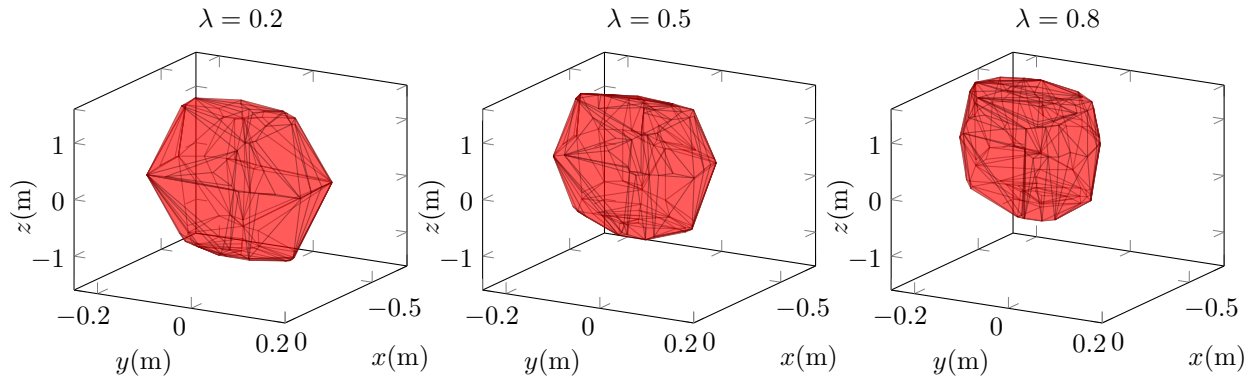


Fig. 11. Morphing between two different robust static stability polyhedrons.

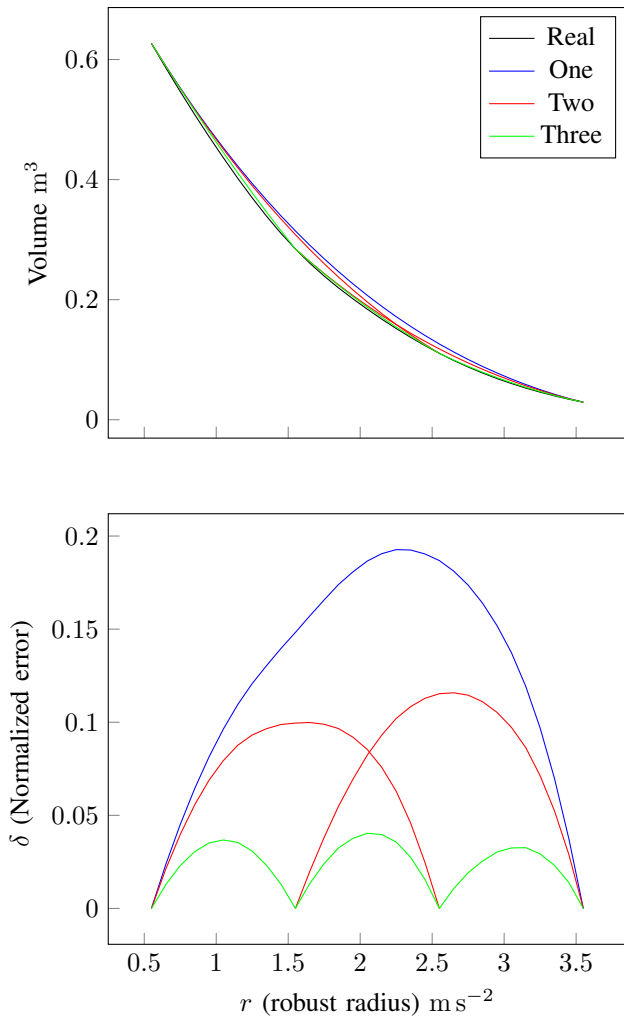


Fig. 12. Comparison of morphing with direct computation for different keypoints.

error remains low (less than 3%) at any margin while the volume error shoots towards 100%.

This confirms that using a 2D approach is equivalent to approximating the 3D polyhedron by a prism formed by its section at some constant height $c_z = h$. This leads to a poor

approximation of the real volume because it is *not* similar to a prism, and this similarity only diminishes with the increase in the stability margin. Note that at any margin the robust 3D polyhedron is *not* included in the prism formed by the “robust polygon” so the “robust prism” is neither a conservative nor optimistic approximation.

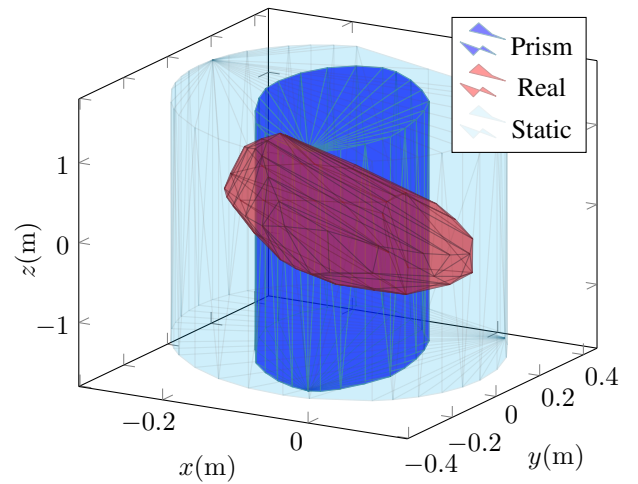


Fig. 13. Comparison between: the statically stable prism in light blue, the robust static polyhedron with a \mathcal{G} generated using $g_m = 2.6 \text{ m s}^{-2}$ in red and the prism formed by the robust polygon at the same margin and height 0.1 m in dark blue. Note that the scale is non-uniform across axes.

IX. CASE STUDY IN MULTI-CONTACT POSTURE GENERATION

We explore how the idea of robust static stability can be applied to posture generation (PG) that aims at finding a statically stable robot stance. We show how we can modify such problems to find *robust* statically stable stances and how this affects the resulting stances.

A. Posture generation

We use the PG framework in [12], [13] and add an extra constraint to maintain the CoM in the robust static stability polyhedron. The core of the PG is to solve a non-linear

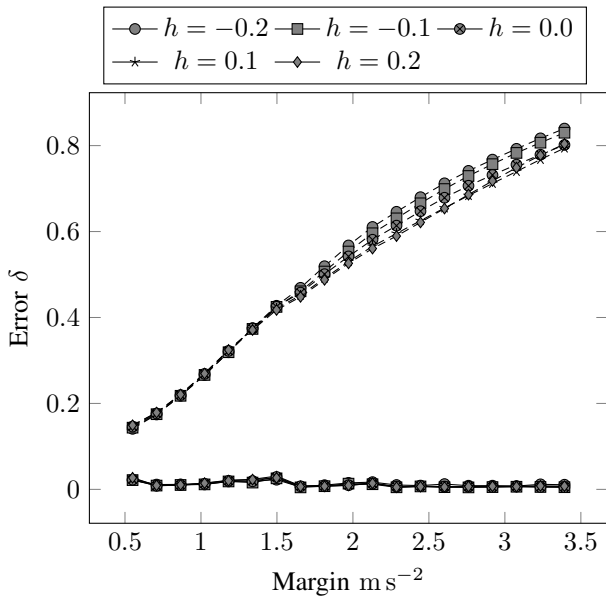


Fig. 14. Compare 2D robust and 3D robust at different heights (in meters). Dashed lines represent volumes error, while solid lines represent areas error.

optimization problem in the generalized robot coordinates q , and the contact forces f . We add our constraint as follows:

$$Hc(q) \leq b \quad (37)$$

Where H and b represent the H-rep of the robust static polyhedron. Other constraints encode:

- Contacts fixed to pre-defined locations;
- Forces in friction cones;
- Both torque and position limits for every joint;
- Self-collision constraints.

The objective function is the distance to a given posture.

It is also possible to extend the state of the posture generator to (q, F) and directly encode the robust stability by specifying that for every $f_i \in F$:

$$A_1(q)f_i + A_2(\tilde{g}_i)c(q) = T(\tilde{g}_i) \quad (38)$$

In this formulation, the contact positions are not fixed: they are a function of q . Thus, they will be determined by the optimization process. However, dimensionality remains a problem in non-linear programming, and extending the state increases drastically the computation times. Therefore, we consider only the effects of adding the constraint eq. (37).

B. Results

We generated three random contact positions: one for each foot and one for the right arm. Then we computed three pairs of postures using:

- Both feet;
- Both feet and the arm;
- The left foot and the arm.

In each pair, the first posture is generated using the static stability constraint while the second one is generated using

the same problem plus the constraint eq. (37). The results are shown in Figure 15. They qualitatively highlight a problem with the regular static stability: the solver has a tendency to stop on the edge of the constraint which is an *unstable* equilibrium position. The postures generated with the robust static stability look more resilient to external perturbations.

Indeed, the main difference is that the CoM is shifted:

- 1) Laterally: away from the edges of the stability region.
- 2) Vertically: the CoM is lower in the robust case.

Simple shrinking of the static stability polygon p would give a similar effect to the first point. Yet, simple isotropic shrinking would fail to account for the specific geometric and frictional properties of each contact. Moreover, this method would not result in CoM altitude changes. Finally, our method allows for explicit acceleration margins whereas shrinking the static stability polyhedron is only an approximation.

This 3D shift in the CoM position also results in a more uniform distribution of contact forces, even though no explicit force objective was formulated.

X. CONCLUSION

We have thoroughly explored the notion of robust static stability polyhedron: its nature and properties, how to compute it and how to use it for testing robust static equilibrium. We have also shown how to approximate variations in the residual radius with polyhedral morphing.

An interesting property is that it is not a right prism: approximating this shape by simply shrinking the static stability polygon is not correct as the height of the CoM plays a role.

As future work, we aim to integrate this computation in our planning and control frameworks. Adding robustness margins will greatly increase the quality of planning and will make our control more resilient to external perturbations. We started by applying this technique to fixed-contacts single posture generation. Yet, the location of the contacts will vary by small increments: it would be paramount to find a way to recompute, in a few milliseconds, a new polyhedron based on the contact shifts. In general, we need to improve the computation time. It would be interesting to exploit the particular characteristics of the problem that may lead to faster computations. Indeed, the high-dimensional polyhedron projection we compute is a cartesian product of cones cut by a 6-dimensional hyperplane, but we only leverage the fact that it is a convex shape. Similarly, it could prove highly beneficial to parallelize some parts of our algorithm, most notably during sampling.

For control, two applications that work at fixed contacts can be considered: control by polyhedron morphing and trajectory planning in polyhedral regions. We already presented control by polygon morphing in [2] and trajectory planning in [1].

Finally, we have shown how to perform iterative sampling, but this opens a new venue of research: we can choose the search direction according to the problem rather than always choosing the optimal direction. This would allow us to only refine (or even only compute) the parts of the robust stability polyhedron that are relevant to the problem being solved.

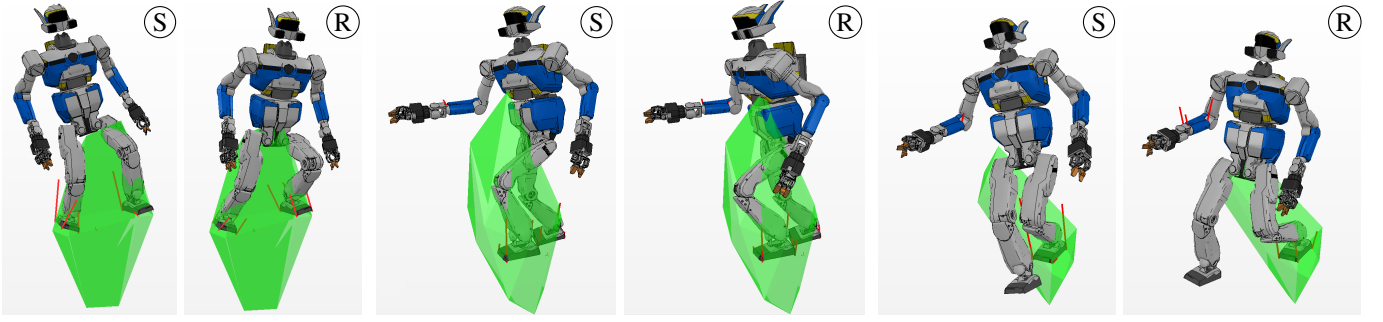


Fig. 15. Results obtained using posture generation. From left to right, both feet, feet and right forearm, left foot and right forearm, each pair is first non-robust, labelled S, then robust, labelled R. The same robust polyhedron is displayed in both images of each pair.

APPENDIX A MATRIX NOTATIONS

We recall the notations in [11] that are introduced to obtain a more compact representation of the Euler-Newton equations. We thus start from them, in the static equilibrium setting:

$$\sum f_i = -mg \quad (39)$$

$$\sum f_i \times (c - r_i) = 0 \quad (40)$$

$$\forall i \quad \|f_i^t\| \leq \mu f_i^n \quad (41)$$

where f_i^t represent the tangential and f_i^n the normal forces. The r_i are the contact positions in the world frame.

We can then slightly reorganize them to isolate the parts that depend on c :

$$\begin{aligned} \sum f_i &= -mg \\ \sum r_i \times f_i - mg \times c &= 0 \\ \forall i \quad \|f_i^t\| &\leq \mu f_i^n \end{aligned} \quad (42)$$

We then define the following quantities:

$$T(g) = \begin{bmatrix} -mg \\ 0_3 \end{bmatrix} \quad (43)$$

$$A_1 = \begin{bmatrix} I_3 & I_3 & I_3 & \cdots & I_3 \\ [r_0]_\times & [r_1]_\times & [r_2]_\times & \cdots & [r_n]_\times \end{bmatrix} \quad (44)$$

$$A_2(g) = \begin{bmatrix} 0 \\ -m[g]_\times \end{bmatrix} \quad (45)$$

$$B_i = I_3 - n_i n_i^T \quad u_i = \mu_i n_i \quad (46)$$

where n_i is the normal at the i^{th} contact, $[\cdot]_\times$ represents the screw symmetric matrix such that $\forall u, v \in \mathbb{R}^3, a \times b = [a]_\times b$. They are such that the system (42) is equivalent to:

$$\begin{aligned} A_1 f + A_2(g)c &= T(g) \\ \forall i \quad \|B_i f_i\| &\leq u_i^T f_i \end{aligned} \quad (47)$$

And thus we can define:

$$B = \begin{bmatrix} B_0 & & & & \\ & B_1 & & & \\ & & \ddots & & \\ & & & \ddots & \\ & & & & B_n \end{bmatrix} \quad u = \begin{bmatrix} u_0 \\ u_1 \\ \vdots \\ u_n \end{bmatrix} \quad (48)$$

Which means that (42) is equivalent to:

$$\begin{aligned} A_1 f + A_2(g)c &= T(g) \\ \|Bf\| &\leq u^T f \end{aligned} \quad (49)$$

APPENDIX B BOUNDS PROOFS

Proposition 2. *If $\exists (l, s) \in \mathbb{R}^2$ such that $\forall f_i$:*

$$u^T f_i \geq u^T f + l \quad (50)$$

$$\|Bf_i\| \leq \|Bf\| + s \quad (51)$$

we have the following implication:

$$\|Bf\| \leq u^T f + l - s \Rightarrow \|Bf_i\| \leq u^T f_i \quad (52)$$

Proof. $\|Bf\| \leq u^T f + l - s$ writes $\|Bf\| + s \leq u^T f + l$ then $\|Bf_i\| \leq \|Bf\| + s \leq u^T f + l \leq u^T f_i$ from eq. (50) and eq. (51). \square

In what follows, we show that bounds l and s exist and subsequently eq. (52) holds.

We need intermediary results that will be used in expressing both l and s .

Lemma 4. *The only possible f_i are given by:*

$$f_i = f + A_1^\dagger (T(\tilde{g}_i) - A_2(\tilde{g}_i)c) + (I - A_1^\dagger A_1)w \quad (53)$$

with \dagger the Moore-Penrose pseudo inverse, and w a vector having the size of contact forces.

Proof. We first exploit the linearity of A_2 (screw operator) and T (stacking acceleration with zero angular momentum):

$$\text{eq. (2) and eq. (4)} \Leftrightarrow A_1 \cdot (f_i - f) + A_2(\tilde{g}_i)c = T(\tilde{g}_i) \quad (54)$$

Since A_1 is full row-rank³, the only possible f_i is given by:

$$f_i = f + A_1^\dagger (T(\tilde{g}_i) - A_2(\tilde{g}_i)c) + (I - A_1^\dagger A_1)w \quad (55)$$

\square

Proof of theorem 1. We develop $u^T f_i$ using Lemma 4, considering the minimal norm solution: $w = 0$.

$$u^T f_i = u^T f + u^T A_1^\dagger (T(\tilde{g}_i) - A_2(\tilde{g}_i)c) \quad (56)$$

$$= u^T f + u^T A_1^\dagger T(\tilde{g}_i) - u^T A_1^\dagger A_2(\tilde{g}_i)c \quad (57)$$

³Except the degenerate unlikely case where all the contact points are aligned.

Now, we use the following properties:

- For any vectors a and b , $|a^T b| \leq \|a\| \|b\|$ (Cauchy-Schwarz inequality);
- $\|u\| = \mu$ (by definition);
- $\|Ax\| \leq \bar{\sigma}(A) \|x\|$ with $\bar{\sigma}(A)$ the largest singular value.

Firstly,

$$u^T A_1^\dagger T(\tilde{g}_i) \geq -\|u\| \|A_1^\dagger T(\tilde{g}_i)\| \quad (58)$$

$$\geq -\mu \bar{\sigma}(A_1^\dagger) m r \quad (59)$$

Secondly,

$$u^T A_1^\dagger A_2(\tilde{g}_i) c \leq \|u\| \|A_1^\dagger A_2(\tilde{g}_i) c\| \quad (60)$$

$$\leq \mu \bar{\sigma}(A_1^\dagger) m r \|c\| \quad (61)$$

$$-u^T A_1^\dagger A_2(\tilde{g}_i) c \geq -\mu \bar{\sigma}(A_1^\dagger) m r \|c\| \quad (62)$$

Thus, from eq. (59) and eq. (62), we have

$$u^T f_i \geq u^T f - \mu r m \bar{\sigma}(A_1^\dagger) (1 + \|c\|) \quad (63)$$

□

Proof of theorem 2. We develop $\|Bf\|$ using Lemma 4, considering the minimal norm solution: $w = 0$.

$$\|Bf_i\| = \|Bf + BA_1^\dagger T(\tilde{g}_i) + BA_1^\dagger A_2(\tilde{g}_i) c\| \quad (64)$$

We first apply the triangular inequality:

$$\|Bf_i\| \leq \|Bf\| + \|BA_1^\dagger T(\tilde{g}_i)\| + \|BA_1^\dagger A_2(\tilde{g}_i) c\| \quad (65)$$

Firstly:

$$\|BA_1^\dagger T(\tilde{g}_i)\| \leq r m \bar{\sigma}(BA_1^\dagger) \quad (66)$$

Secondly:

$$\|BA_1^\dagger A_2(\tilde{g}_i) c\| \leq r m \bar{\sigma}(BA_1^\dagger) \|c\| \quad (67)$$

We thus have:

$$\|Bf_i\| - \|Bf\| \leq r m \bar{\sigma}(BA_1^\dagger) (1 + \|c\|) \quad (68)$$

□

We thus see that tightness can only be achieved if the following conditions are achieved at the same time:

- $\|Aa\| \leq \bar{\sigma} \|a\|$ is tight whenever a is aligned with the singular vector associated with the maximum singular value, say \bar{v} . This gives us:
 - $\tilde{g}_i = r \bar{v}$ thus \tilde{g}_i is parallel to \bar{v} .
 - $A_2(\tilde{g}_i) c = k \bar{v}$ i.e. $\tilde{g}_i \times c = \alpha r \|c\| \bar{v}$ that is to say a vector perpendicular to \tilde{g}_i is parallel to \bar{v} .
- $\|A_2(\tilde{g}_i) c\| \leq m r \|c\|$ is equal iff $c \perp \tilde{g}_i$

Which are equivalent to those presented in section III-B.

APPENDIX C

CONVERGENCE PROPERTIES

In section VI, we state that Algorithm 1 has a linear or quasi-linear rate of convergence. Indeed, from intuition alone, approximating a circle by a square only requires 4 vertices, but approximating a sphere by a cube requires 8 vertices. As both Algorithm 1 and [11] generate vertices one by one, we expected that the augmentation of the dimension would slow down the rate of convergence.

This is reinforced by the stochastic geometry literature: it is proven in [24], [8] that the rate of convergence depends on the dimension as follows. Consider \mathcal{P} the shape to be approximated, $\mathcal{P}_{\text{outer}}^n$ and $\mathcal{P}_{\text{inner}}^n$ the best circumscribing and inscribed polyhedral approximations of \mathcal{P} respectively. $\mathcal{P}_{\text{outer}}^n$ has n facets tangent to \mathcal{P} while $\mathcal{P}_{\text{inner}}^n$ has n vertices located on the surface $\partial\mathcal{P}$ of \mathcal{P} . If \mathcal{P} is convex (gaussian curvature positive or null everywhere) and \mathcal{C}^2 -smooth, then for the symmetric difference metric δ the best approximations verify:

$$\delta(\mathcal{P}_{\text{outer}}^n, \mathcal{P}) \sim \frac{1}{2} \text{del}_{N-1} \mathcal{A}(\partial\mathcal{P}) \frac{N+1}{N-1} \frac{1}{n^{\frac{2}{N-1}}} \quad (69)$$

$$\delta(\mathcal{P}, \mathcal{P}_{\text{inner}}^n) \sim \frac{1}{2} \text{div}_{N-1} \mathcal{A}(\partial\mathcal{P}) \frac{N+1}{N-1} \frac{1}{n^{\frac{2}{N-1}}} \quad (70)$$

with del and div being constants that only depend on the dimension N . Note that \mathcal{A} in this context denotes the affine surface area rather the regular surface area. The result can be extended to the best approximating *sequences* of \mathcal{P} .

Other similar results hold for different metrics, the most interesting being the Schneider metric. The Schneider metric $\delta^{\text{SCH}}(\mathcal{P}_1, \mathcal{P}_2)$ only applies if $\mathcal{P}_2 \subset \mathcal{P}_1$. It is the maximum volume of caps of \mathcal{P}_1 cut by the supporting hyperplanes of \mathcal{P}_2 . In Algorithm 1 δ^{SCH} is implicitly used: we choose the search direction perpendicular to the facet that forms the biggest uncertainty volume i.e. maximizes $\delta^{\text{SCH}}(\mathcal{P}_{\text{outer}}^n, \mathcal{P}_{\text{inner}}^n)$.

Our measure of convergence is defined by:

$$\delta(\mathcal{P}_{\text{outer}}^n, \mathcal{P}_{\text{inner}}^n) = \delta(\mathcal{P}_{\text{outer}}^n, \mathcal{P}) + \delta(\mathcal{P}, \mathcal{P}_{\text{inner}}^n) \quad (71)$$

Thus, the previous results entail:

- Our algorithm, if it is indeed linear, is optimal in terms of rate of convergence.
- Approximating the sphere in section VI is the worst-case scenario as it maximizes $\mathcal{A}(\partial\mathcal{P})$ for a given volume.

However, we cannot directly use this result to prove our rate of convergence:

- It only applies to \mathcal{C}^2 bodies, while the volume we are approximating is \mathcal{C}^2 almost-everywhere only: it has straight edges.
- We are computing two sequences of approximations of \mathcal{P} , that minimize δ^{SCH} at each step but they probably do not minimize δ .
- This result is only asymptotic: as we stop the algorithm after a finite (potentially small) number of iterations, we need to prove a result that holds even for small n .

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