

Distance-sensitive planar point location

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Distance-Sensitive Planar Point Location $\stackrel{\leftrightarrow}{\approx}$

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Abstract

Let S be a connected planar polygonal subdivision with n edges that we want to preprocess for point-location queries, and where we are given the probability γ_i that the query point lies in a polygon P_i of S. We show how to preprocess S such that the query time for a point $p \in P_i$ depends on γ_i and, in addition, on the distance from p to the boundary of P_i —the further away from the boundary, the faster the query. More precisely, we show that a point-location query can be answered in time $O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$, where Δ_p is the shortest Euclidean distance of the query point p to the boundary of P_i . Our structure uses O(n) space and $O(n \log n)$ preprocessing time. It is based on a decomposition of the regions of S into convex quadrilaterals and triangles with the following property: for any point $p \in P_i$, the quadrilateral or triangle containing p has area $\Omega(\Delta_p^2)$. For the special case where S is a subdivision of the unit square and $\gamma_i = \operatorname{area}(P_i)$, we present a simpler solution that achieves a query time of $O\left(\min\left(\log n, \log \frac{1}{\Delta_p^2}\right)\right)$. The latter solution can be extended to convex subdivisions in three dimensions.

Keywords: point location, quadtree, mesh generation

1. Introduction

Point location is one of the most fundamental problems in computational geometry. Given a subdivision \mathcal{S} the goal is to preprocess it so that we can

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determine efficiently which region of S contains a query point p. Many different variants of the point-location problem exist; in our work we first focus on planar point location in polygonal subdivisions and later extend one of our results to convex polyhedral subdivisions in three dimensions. In the following, unless otherwise specified, our subdivision S is subdivision of a polygonal domain in the plane into polygons. The subdivision need not be conforming—we may have Tjunctions, for instance—but when considering a polygon P_i of the subdivision we ignore the subdivision vertices whose angle inside P_i is exactly π . A triangulation is a subdivision consisting of triangles.

There are several different solutions for planar point location that are worstcase optimal. In particular, there are structures that require $O(n \log n)$ preprocessing, use O(n) space, and can answer a point-location query in $O(\log n)$ time; see the surveys by Preparata [18] and Snoeyink [22] for an overview. In three dimensions no point location structure is known for general subdivisions that uses linear space and has logarithmic query time. For convex subdivisions Preparata and Tamassia [19] showed that combining dynamic planar point location and persistency techniques yield an $O(n \log^2 n)$ space structure that answers queries in $O(\log^2 n)$ time. This method was later extended and improved so that it works for general subdivisions and requires only $O(n \log n)$ space and preprocessing time for $O(\log^2 n)$ query time [12, 22].

For planar point location a query time of $O(\log n)$ is optimal in the worst case, but it may be possible to do better for certain types of query points. For example, if the query points are not distributed uniformly among the regions of S, then it may be desirable to reduce the query time for points in frequently queried regions. Iacono [13] showed that this is indeed possible: given a triangulation S where each triangular region R_i has a probability γ_i associated with it—the probability that the query point p falls in R_i —then one can answer a point-location query in expected time O(H(S)), where

$$H(\mathcal{S}) := \sum_{R_i \in \mathcal{S}} \gamma_i \log \frac{1}{\gamma_i},$$

is the entropy of S. This result is optimal, because the entropy is a lower bound on the expected query time [16, 21]. Several other point-location structures have been proposed that answer queries in O(H(S)) expected time [1, 2]. The structure presented by Arya, Malamatos, and Mount [1] is relatively simple and efficient in practice. It works for subdivisions with constant-complexity regions and, for any region R_i the worst-case query time for points inside R_i is $O(1 + \min(\log \frac{1}{\gamma_i}, \log n))$. The results mentioned so far assume that the distribution is known in advance. Recently Iacono [14] proposed an algorithm that eventually achieves O(H(S)) query time, but does not need any knowledge of the query distribution. Instead, the algorithm changes the structure according to the queries received. The results mentioned above require the regions of the input subdivision S to have constant complexity. This requirement is necessary. Indeed, if a subdivision with n edges has only two regions, each with associated probability 1/2, then we cannot hope to achieve O(1) query time. One could of course subdivide the regions into constant complexity regions, say triangles, and distribute the query probability evenly among these regions. However, in many cases one would expect that queries are not evenly distributed within each polygon. For example if queries come from users selecting polygons by clicking on them, one would expect most queries to occur far from the boundary as users are inclined to click in the 'middle' of a region. This raises the question if it is possible to improve query times depending on where the query point is within the polygon that contains it. In our work we investigate the possibility of relating the query time to the distance of a query point to the nearest point on the boundary of the region that contains it. We call this *distance-sensitive point location*.

Differentiating between query points within higher complexity polygons is not new. Collette *et al.* [7] showed how to compute, for any simple polygon P and any probability distribution over P, a Steiner triangulation with near-optimal entropy, and they proved that the minimum entropy of any triangulation is a lower bound on the expected query time for point-location in the linear decisiontree model. By applying their Steiner triangulation to every region in the given subdivision, and using the resulting triangles as input for an entropy-based point-location structure, near-optimal expected query time is achieved. In the case of distance-sensitive point location we could define a probability distribution based on the distance of points to the region boundary and construct such a Steiner triangulation. Unfortunately, a near-optimal entropy does not imply any bounds on specific query points. Indeed the construction by Collette et al. can generate very small triangles, even in high probability areas. A point p that is far from the boundary can end up in such a very small triangle, which has a small total probability. As a result a query for p has a long query time. We will focus on creating a point location structure that guarantees fast query time for any point far from the region boundary.

Problem definition. Let p be a query point inside polygon $P_i \in S$ with area $\operatorname{area}(P_i)$ and probability γ_i that a query point is inside P_i . We want the time of a query for p to be

$$O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$$

Here, Δ_p denotes the minimum Euclidean distance from p to the boundary of P_i . When the polygons in the subdivision have constant complexity, then this can be achieved using, for instance, the entropy-based point-location structure of Arya, Malamatos, and Mount [1]. Since for any point $p \in P_i$ the distance to the boundary of P_i is $O(\sqrt{\operatorname{area}(P_i)})$, this gives the desired query bound. When polygons have higher complexity we can also use an entropy-based structure, but first have to decompose each polygon into constant complexity regions and assign probabilities appropriately. Specifically we show that it is sufficient to compute for each polygon $P \in S$ with n_P vertices a distance-sensitive decomposition of P into $O(n_P)$ regions with the following properties:

- each region R is convex and has constant complexity;
- for some absolute constant α the decomposition has the α -distance property: for any point $p \in P$, the region R containing p has area at least $\alpha \cdot \Delta_p^2$, where Δ_p is the distance from p to the boundary of P.

The entropy-based search structure by Arya, Malamatos, and Mount [1], which will serve as the backbone of our algorithm, requires its input regions to have constant complexity. Here the complexity of a region is counted as explained earlier: if the interior angle within a region P_i at a subdivision vertex is exactly π , then that vertex does not count towards the complexity of P_i . For instance, the shaded region in the figure on the right has only five vertices.

The problem of computing a decomposition with these properties can be considered a mesh-generation problem. Many different types of meshes exist; see the survey by Bern [3] for an overview. In several of these meshes the number of mesh elements is linear in the complexity of the polygon, and the mesh elements are "well-shaped". For example, the the meshing algorithm proposed by Bern *et al.* [5] produces triangles with angles of at most 90°. There are also meshes that are designed to be more detailed near the polygon boundary and coarser further away from the boundary. These meshes, however, do not guarantee a relation between the distance to the boundary and the size of mesh elements [5, 20] or they do not have a bound on the number of mesh elements [4]. To the best of our knowledge no published mesh generation method guarantees that the mesh consists of $O(n_P)$ elements that have the required distance property.

Another "query-sensitive" subdivision was defined by Mitchel, Mount, and Suri to faciltate ray shooting [17]. It has the property that the cost of shooting a ray (i.e., walking along it through the subdivision, from its origin until the first point of intersection with the obstacle) is proportional to its "cover complexity," which, roughly speaking, is the minimum number of disks that are required to cover this portion of the ray, with each disk not intersecting "too much" of an obstacle. In our application, the role of the obstacle is taken by the exterior of the region to be subdivided. However, the structure of [17] does not seem have the right properties for our purposes.

Our results. We start by describing in more detail how a distance-sensitive decomposition can be used to construct a distance-sensitive point-location structure. We then continue by giving algorithms to compute distance-sensitive decompositions. For convex polygons we actually do not need to use non-conforming subdivisions: we show that any convex polygon can be triangulated in such a way that the resulting triangulation has the α -distance property for $\alpha = 1$. For possibly non-convex simple polygons we investigate several different settings that have different restrictions on the resulting decomposition. We show that it is not always possible to create a conforming triangulation with the α -distance property without using Steiner points, and that the number of Steiner points needed in such a triangulation cannot be bounded as a function of the complexity of the polygon P. Instead, we show that any simple polygon P can be decomposed into $O(n_P)$ non-conforming convex quadrilaterals and triangles that have the α -distance property for some absolute constant $\alpha > 0$. The decomposition can be computed in $O(n_P \log n_P)$ time. This result is used to obtain a linear-size data structure for point location in a planar connected polygonal subdivision S, such that the query time for a point p in a polygon $P_i \in S$ is $O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$, where Δ_p is the distance from the query point p to the boundary of its containing region.

Lastly we investigate a special case in which the query bound is based only on the distance of a query point to the boundary. Specifically, assuming the subdivision is contained in a square of area 1, we present a data structure that achieves a query time of $O\left(\min\left(\log n, 1 + \log\frac{1}{\Delta_p^2}\right)\right)$ for a point p. The new structure is based on a depth-bounded quadtree and a worst-case optimal point-location structure, both of which can be constructed in $O(n \log n)$ time and O(n) space. The more general structure presented above achieves the same bounds if we choose $\gamma_i = \operatorname{area}(P_i)$, but we believe the new structure is much simpler and may be faster in practice. As a bonus, the new structure achieves the more general bound of $O\left(\min\left(\log n, 1 + \log\frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$ for any subdivision of the unit square where $\gamma_i = O(\operatorname{area}(P_i))$.

The simpler structure also extends to three dimensions. Specifically, given a convex polyhedral subdivision contained in a unit cube with n edges, we show how to construct a distance-sensitive point location structure in $O(n \log^2 n)$ time and $O(n \log n)$ space that answers a query for a point p in $O(1 + \log \frac{1}{\Delta_p^2})$ time if $\Delta_p \geq \sqrt{3}/\sqrt[3]{n}$ and $O(\log^2 n)$ time otherwise. Note that the space requirement comes from the worst-case point location structure and not the additional octree structure that allows for distance-sensitive queries.

2. Distance-sensitive decomposition of simple polygons

As argued in the introduction we can use entropy-based point location structures to create a distance-sensitive point location structure by first creating a distance-sensitive decomposition for the polygons of the input subdivision. To avoid confusion we use the term *polygon* for polygons of the input subdivision S and *region* for the regions of the decomposition of a polygon. Recall that we define a distance-sensitive decomposition as follows: Let P be a simple polygon with n_P edges. A distance-sensitive decomposition of P consists of $O(n_P)$ regions with the following properties:

- each region R is convex and has constant complexity;
- for some absolute constant α the decomposition has the α -distance property: for any point $p \in P$, the region R containing p has area at least $\alpha \cdot \Delta_p^2$, where Δ_p is the Euclidean distance from p to the boundary of P.

Given a subdivision S and for each polygon $P_i \in S$ its distance-sensitive decomposition $\mathcal{P}_i^{\text{dec}}$ we can assign each region $R \in \mathcal{P}_i^{\text{dec}}$ a weight $\gamma_i \cdot \operatorname{area}(R)/\operatorname{area}(P_i)$.

We then build the entropy-based structure by Arya *et al.* [2] on the union of the distance-sensitive decompositions of all polygons in S using the weights for the probability distribution. Now a point p with distance Δ_p to the nearest boundary of the subdivision must be contained in a region R with weight $\gamma_i \cdot \operatorname{area}(R)/\operatorname{area}(P_i) \geq \alpha \cdot \gamma_i \cdot \Delta_p^2/\operatorname{area}(P_i)$. It follows that the query time for pis

$$O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right).$$

So once we have a distance-sensitive decomposition it is easy to construct a distance-sensitive point location structure.

Theorem 1. Let S be a subdivision, where for each polygon $P_i \in S$ we are given a distance-sensitive decomposition $\mathcal{P}_i^{\text{dec}}$. Then we can construct in $O(n \log n)$ expected time a point location for S such that, for any query point p, the query time is $O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$, where Δ_p is the distance from p to the boundary of the region containing p.

Note that the expectation in the construction time has nothing to do with the probabilities γ_i , but it is because Arya *et al.* use randomized incremental construction to build their data structure. Also note that the distance-sensitive decomposition may be non-conforming, that is, the boundary-edges of a region may contain many interior vertices that are not counted towards its complexity. Indeed, since Arya *et al.* use randomized incremental insertion of maximal segments to build their structure, it is not a problem if the decomposition is non-conforming. In the remainder of this section we focus on constructing distance-sensitive decompositions, first for convex polygons and then for arbitrary simple polygons.

2.1. Convex polygons

As a warm-up exercise, we start with the problem of decomposing a convex polygon P with n_P vertices so that the decomposition has the α -distance property for $\alpha = 1$. For this case the decomposition is actually a triangulation.

Our algorithm is quite simple. First we split P by adding a diagonal between the vertices defining the diameter of P. We further decompose each of the two resulting subpolygons using a recursive algorithm, which we describe next. We call the edges of the input polygon P polygon edges and the edges created by the subdivision process subdivision edges. The boundary of each subpolygon we recurse on consists of one subdivision edge and a convex chain of polygon edges, where the angles between the chain and the subdivision edge are acute. Let Qbe such a subpolygon and e the corresponding subdivision edge. We construct the largest area triangle T contained in Q that has e as an edge by finding the vertex v on the convex chain that is farthest from e. Because the chain is convex this vertex can be found in $O(\log n_Q)$ time, where n_Q is the number of vertices of Q.

Theorem 2. For any convex polygon P with n_P vertices we can compute in $O(n_P \log n_P)$ time a triangulation that has the 1-distance property.

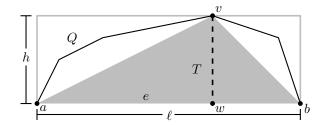


Figure 1: A triangle in polygon Q is defined by the subdivision edge e and the point q furthest away from e.

Proof. Consider the algorithm described above. We first show that the two angles between the subdivision edge e and the convex chain inside a subpolygon Q are acute by showing that |e| is the diameter of Q. For the first two subpolygons, created by cutting the convex polygon across the diagonal the length of the subdivision edge is the diameter of the subpolygon by definition. Now consider a subpolygon Q with subdivision edge e and triangle T formed by e = (a, b) and the furthest point $v \in Q$ from e, see Figure 1. This creates up to two new subpolygons Q_1 and Q_2 with $e_1 = (a, v)$ and $e_2 = (v, b)$ as subdivision edges. Since the angles between e and the convex chain are acute it follows that Q is contained in a rectangle with side length e and height h = dist(v, e). To prove that $|e_1|$ and $|e_2|$ are the diameters for Q_1 and Q_2 respectively consider a point $w \in e$ that is closest to v. The edge (v, w) divides the rectangle into two rectangles R_1 containing Q_1 and R_2 containing Q_2 . The edges e_1 and e_2 are the diameters of these rectangles and it follows that they are also the diameters of Q_1 and Q_2 .

Now consider a subpolygon Q with base edge e and a furthest point $v \in Q$ from e. Since |e| is the diameter Q, the angles at e's endpoints are acute and Q must be contained in an $\ell \times h$ rectangle where $\ell = |e|$ and $h = \operatorname{dist}(v, e)$. It follows that for any point $p \in T$ we have

$$\Delta_p^2 \le \min(h, \ell/2)^2 \le h\ell/2 = \operatorname{area}(T).$$

The diameter of a convex polygon can be computed in $O(n_P)$ time, and the creation of each triangle takes $O(\log n_P)$ time. Since there are $n_P - 2$ triangles it follows that the algorithm takes $O(n_P \log n_P)$ time.

Combining this result with Theorem 1 we obtain the following corollary.

Corollary 3. Let S denote a convex planar polygonal subdivision with O(n) vertices and let γ_i for each $P_i \in S$ denote the probability that a query point lies in P_i . We can construct in $O(n \log n)$ expected time a point location structure that uses O(n) space and answers a query with a point p in $O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$ time, where Δ_p denotes the Euclidean distance from p to the nearest point on any edge of S.

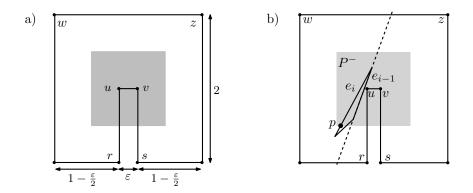


Figure 2: a) Any triangulation of P with the α -distance property requires many Steiner points. b) Triangle T_{i-1} intersects the boundary of P^- in p.

2.2. Arbitrary polygons

We now consider non-convex polygons. We wish to compute a decomposition of a simple polygon P into constant-complexity regions that have the α -distance property. This is not always possible with a triangulation. Consider the polygon in Figure 2, where the width of the middle column coming up from the bottom edge is ε . Any triangulation of the polygon in Figure 2a must include triangle uvw or uvz, and when ε tends to zero the α -distance property is violated for points in the middle of these triangles. A Steiner triangulation with the α distance property always exists: the quadtree-based mesh of Bern *et al.* [4] can be adapted to have the α -distance property—the (small) adaptations are required only around acute angles. However, the number of Steiner points and, hence, the number of triangles cannot be bounded as a function of the number of vertices of P. Next we show that this is necessarily so.

Theorem 4. For any constant $\alpha > 0$ and any m > 0, there is a simple polygon P with eight vertices such that any Steiner triangulation of P with the α -distance property uses at least m Steiner points.

Proof. Let P be the polygon shown in Figure 2a. Consider a Steiner triangulation \mathcal{T} of P with the α -distance property. Let T_0 be the triangle in \mathcal{T} that has uv as an edge.¹ If ε is very small then the other two edges of T_0 cannot be very long either, otherwise the α -distance property is violated inside T_0 . This in turn implies that the neighboring triangles of T_0 cannot be very large. The idea is to repeat this argument to show that many triangles are needed to cover P.

Specifically, we define a sequence of triangles T_0, T_1, T_2, \ldots , as follows. Suppose we are given a triangle T_i and an edge e_i bounding T_i from below. (For

¹The edge uv can contain Steiner vertices in its interior, as the only requirement we have for the Steiner triangulation is that any two triangles either meet in a single vertex, along a complete edge, or not at all. When uv contains Steiner vertices, we can replace uv by any subedge of uv, and the argument still holds.

i = 0, we have $e_i = uv$.) Consider the other two edges of T_i . We select one of these two edges as e_{i+1} and define T_{i+1} as the triangle directly above e_{i+1} . We select e_{i+1} as follows. If only one of the edges bounds T_i from above, then this edge is selected. If both edges bound T_i from above, then we select the edge with the smaller absolute slope. This selection guarantees that for every edge e_i at least one endpoint is above e_0 .

Our goal is now to prove that the size of the triangles T_0, T_1, \ldots does not increase too rapidly—more precisely, that T_{i+1} cannot be arbitrarily larger than T_i . This requires an invariant on the length of the edges e_i , but also on their absolute slope. We denote the absolute slope of e_i by σ_i . Thus $\sigma_i = |e_i|_y/|e_i|_x$, where $|e_i|_x$ and $|e_i|_y$ denote the lengths of the projection of e_i on the x- and y-axis. Let P^- denote the square with edge length 1 centered at the midpoint of uv. In Figure 2a this square is shaded. Our argument will use the fact that for T_i inside P^- the nearest boundary point for any $p \in T_i$ lies on uv, ur, or vs. We show that both the slope and length of edge e_i are bounded as a function of i, and that e_i remains inside P^- , until $\sigma_i \cdot |e_i|$ is large enough. More precisely, we can prove that as long as $\max(4, \sigma_i^2) \cdot |e_i| < \frac{\alpha}{8\sqrt{2}}$ the following three properties hold, where (i) and (ii) are needed to prove (iii):

- (i) edge e_i is contained in P^- ;
- (ii) the slope σ_i of e_i satisfies $\sigma_i \leq (2^{i+1}-2)/\alpha$; (iii) edge e_i has length at most $8\varepsilon \cdot 2^{(i+1)(i+7)}/(\alpha^{3i})$.

These properties can be proven using induction, where the proof for (ii) requires (i) and the proof for (iii) requires (i) and (ii). It is easy to see that (i), (ii) and (iii) are true for e_0 and the step cases are given in Lemmas 5, 6 and 7, respectively. It follows from property (iii) that we can always choose ε small enough that we need at least m Steiner points before T_i can leave the square P^- .

Lemma 5. If for e_j with $0 \le j < i$ we have $\max(4, \sigma_j^2) \cdot |e_j| < \frac{\alpha}{8\sqrt{2}}$ and e_j is contained in P^- , then e_i is contained in P^- .

Proof. We assume for a contradiction that e_i extends outside of P^- and show that if this is the case, then T_{i-1} does not have the α -distance property for the given α . The area of T_{i-1} is upper bounded by $|e_{i-1}| \cdot |e_i| \leq |e_{i-1}| \cdot 2\sqrt{2}$. Since e_i extends outside of P^- and e_{i-1} is inside it there must be a point $p \in T_{i-1}$ that is on the boundary of P^- . If p is on the left, top or right edge of P^- then $\Delta_p \geq (1-\varepsilon)/2 \geq 1/4$. If p is on the bottom edge of P^- we can use the slope of e_{i-1} and the fact that its top endpoint is above e_0 to bound the distance from p to the boundary of P. Without loss of generality assume that p is to the left of ur. Since one endpoint of e_{i-1} is above e_0 and e_{i-1} cannot intersect e_0 the distance from p to ur (the nearest boundary edge) is at least $1/(2\sigma_{i-1})$, see also Figure 2b. We get that $\Delta_p \geq 1/(2 \max(2, \sigma_{i-1}))$. This would imply that

$$\operatorname{area}(T_i) \le |e_{i-1}| \cdot 2\sqrt{2} < \frac{\alpha}{8\sqrt{2}\max(4,\sigma_{i-1}^2)} \cdot 2\sqrt{2} \le \alpha \cdot \Delta_p^2,$$

contradicting that T_i has the α -distance property. Hence, we can conclude that e_i must be contained in P^- .

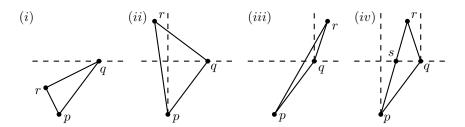


Figure 3: The four cases for a point r used in the proof of Lemma 6

Lemma 6. If for all $0 \le j < i$ the edge e_j is inside P^- and $\max(4, \sigma_j^2) \cdot |e_j| < \frac{\alpha}{8\sqrt{2}}$, then $\sigma_i \le (2^{i+1}-2)/\alpha$.

Proof. First note that from Lemma 5 we know that e_i is contained in P^- . Let p and q denote the bottom and top endpoints of e_{i-1} , and let r denote the third vertex of T_{i-1} . Without loss of generality we assume that p is to the left of q. We distinguish cases based on the location of r relative to p and q (see Figure 3).

Case (i): r is below q. If r is below q, then e_i is the edge qr. Since r must be above the supporting line of pq and left of q, we get that $\sigma_i \leq \sigma_{i-1}$.

Case (ii): r is to the left of p and above q. If r is left of p and above q, then $e_i = rq$. Let $\operatorname{area}(\mathbf{T}_{i-1})$ denote the area of \mathbf{T}_{i-1} and $\operatorname{dist}(\mathbf{T}_{i-1})$ the maximum distance from any point in \mathbf{T}_{i-1} to the boundary of P. Since \mathbf{T}_{i-1} is contained within P^- and q is above e_0 , it follows that $\operatorname{dist}(\mathbf{T}_{i-1}) \geq |rq|_y$. Due to our assumptions on the positions of p, q, r it follows that $\operatorname{area}(\mathbf{T}_{i-1}) \leq |rp|_y \cdot |rq|_x$. This allows us to bound the slope σ_i as

$$\begin{aligned} \sigma_i - \sigma_{i-1} &= \sigma_i \left(1 - \frac{|pq|_y}{|pq|_x} \cdot \frac{|rq|_x}{|rq|_y} \right) \le \sigma_i \left(1 - \frac{|pq|_y}{|rq|_y} \right) \le \sigma_i \left(1 - \frac{|pq|_y}{|rp|_y} \right) \\ &= \sigma_i \frac{|rp|_y - |pq|_y}{|rp|_y} = \sigma_i \frac{|rq|_y}{|rp|_y} = \frac{|rq|_y^2}{|rq|_x \cdot |rp|_y} \le \frac{\operatorname{dist}(\mathbf{T}_{i-1})^2}{\operatorname{area}(\mathbf{T}_{i-1})} \le 1/\alpha. \end{aligned}$$

Case (iii): r is to the right of q. As before we have $dist(T_{i-1}) \ge |rq|_y$ and $area(T_{i-1}) \le |rq|_y \cdot |pq|_x$. We get

$$\sigma_i - \sigma_{i-1} = \frac{|rp|_y}{|rp|_x} - \frac{|pq|_y}{|pq|_x} \le \frac{|rp|_y - |pq|_y}{|pq|_x} = \frac{|rq|_y}{|pq|_x} = \frac{|rq|_y^2}{|rq|_y \cdot |pq|_x} \le 1/\alpha.$$

Case (iv): r is horizontally between p and q. This case provides us with two edges that face upward. Recall that in this case e_i is the edge with smaller slope. We further split this case into three subcases. First assume that $\sigma_{rp} \leq \sigma_{rq}$. Let s denote a point on rp with the same y-coordinate as q, so $|sq|_x = |rp|_x \cdot \frac{|rq|_y}{|rp|_y} + |rq|_x$. We bound area $(T_{i-1}) \leq |sq|_x \cdot |rp|_y$ and $dist(T_{i-1}) \geq |rq|_y$. We get

$$\sigma_i - 2\sigma_{i-1} = \sigma_{rp} - 2\sigma_{pq} = \sigma_{rp} \left(1 - 2\frac{|pq|_y}{|pq|_x} \cdot \frac{|rp|_x}{|rp|_y} \right) \le 2\sigma_{rp} \left(1 - \frac{|pq|_y}{|rp|_y} \right)$$

$$=\sigma_{rp}\frac{|rq|_{y}}{|rp|_{y}} = 2\frac{|rq|_{y}}{\frac{2}{\sigma_{rp}}|rp|_{y}} \le 2\frac{|rq|_{y}}{\left(\frac{1}{\sigma_{rp}} + \frac{1}{\sigma_{rq}}\right) \cdot |rp|_{y}} = 2\frac{|rq|_{y}}{|rp|_{y} \cdot \left(\frac{|rp|_{x}}{|rp|_{y}} + \frac{|rq|_{x}}{|rq|_{y}}\right)} \\ = 2\frac{|rq|_{y}^{2}}{|rp|_{y} \cdot \left(|rp|_{x} \cdot \frac{|rq|_{y}}{|rp|_{y}} + |rq|_{x}\right)} = 2\frac{|rq|_{y}^{2}}{|rp|_{y} \cdot |sq|_{x}} \le 2\frac{\operatorname{dist}(\mathrm{T}_{i-1})^{2}}{\operatorname{area}(\mathrm{T}_{i-1})} \le 2/\alpha.$$

The second case we assume that $\sigma_{rq} < \sigma_{rp}$ and $|rp|_x \ge |pq|_x/2$. This leads to a very similar calculation to the previous case, namely

$$\begin{aligned} \sigma_{i} - 2\sigma_{i-1} &= \sigma_{rq} - 2\sigma_{pq} = \sigma_{rq} \left(1 - 2\frac{|pq|_{y}}{|pq|_{x}} \cdot \frac{1}{\sigma_{rq}} \right) \leq \sigma_{rq} \left(1 - 2\frac{|pq|_{y}}{|pq|_{x}} \cdot \frac{1}{\sigma_{rp}} \right) \\ &\leq \sigma_{rq} \left(1 - \frac{|pq|_{y}}{|rp|_{y}} \right) = \sigma_{rq} \frac{|rq|_{y}}{|rp|_{y}} = 2\frac{|rq|_{y}}{\frac{2}{\sigma_{rq}}} |rp|_{y} \leq 2\frac{|rq|_{y}}{\left(\frac{1}{\sigma_{rp}} + \frac{1}{\sigma_{rq}}\right) \cdot |rp|_{y}} \\ &= 2\frac{|rq|_{y}}{|rp|_{y} \left(\frac{|rp|_{x}}{|rp|_{y}} + \frac{|rq|_{x}}{|rq|_{y}}\right)} = 2\frac{|rq|_{y}}{|rp|_{y} \left(|rp|_{x} \cdot \frac{|rq|_{y}}{|rp|_{y}} + |rq|_{x}\right)} = 2\frac{|rq|_{y}^{2}}{|rp|_{y} \cdot |sq|_{x}} \leq 2/\alpha. \end{aligned}$$

Lastly we assume that $\sigma_{rp} < \sigma_{rp}$ and $|rq|_x \ge |pq|_x/2$. Here we use slightly different bounds on the area, namely that $\operatorname{area}(\mathbf{T}_{i-1}) \le |rp|_y \cdot |pq|_x$. We still have the bound of $\operatorname{dist}(\mathbf{T}_{i-1}) \ge |rq|_y$ on the distance to the boundary, which gives us

$$\begin{aligned} \sigma_i - 2\sigma_{i-1} &= \sigma_{rq} - 2\sigma_{pq} = \sigma_{rq} \left(1 - 2\frac{|pq|_y}{|pq|_x} \cdot \frac{|rq|_x}{|rq|_y} \right) \le \sigma_{rq} \left(1 - \frac{|pq|_y}{|rq|_y} \right) \\ &\le \sigma_{rq} \left(1 - \frac{|pq|_y}{|rp|_y} \right) = \sigma_{rq} \frac{|rp|_y - |pq|_y}{|rp|_y} = \frac{|rq|_y^2}{|rq|_x \cdot |rp|_x} \le 2\frac{|rq|_y^2}{|pq|_x \cdot |rp|_x} \le 2/\alpha. \end{aligned}$$

In each case we find that

$$\sigma_i \le 2\sigma_{i-1} + 2/\alpha \le 2 \cdot (2^i - 2)/\alpha + 2/\alpha = (2^{i+1} - 2)/\alpha.$$

Lemma 7. If for all j < i the edge e_j is inside P^- and $\max(4, \sigma_j^2) \cdot |e_j| < \frac{\alpha}{8\sqrt{2}}$, then $|e_i| \leq 8\varepsilon \cdot 2^{(i+1)(i+7)}/\alpha^{3i}$.

Proof. Note that from Lemmas 5 and 6 we already know that e_i is inside $P^$ and $\sigma_i \leq (2^{i+1}-2)/\alpha$. We first give bounds on the area and the distance to the boundary of T_{i-1} . We upper bound $\operatorname{area}(T_{i-1}) \leq |e_{i-1}| \cdot |e_i|$. For a lower bound on the distance to the boundary we look at the distance of points on e_i to the boundary. From property (i) in Theorem 4 we know that e_i is also inside

²This step assumes that $\frac{|rp|_x}{|pq|_x} \ge 1/2$, which follows from our assumption that $\sigma_{rp} \le \sigma_{rq}$.

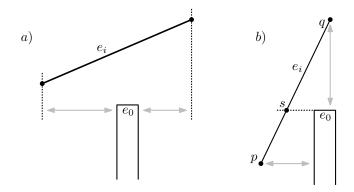


Figure 4: Illustration of the two cases in the proof of Lemma 7 based on the slope of e_i . Gray double arrows indicate distances used in the proof.

 P^- . Let p and q denote the two endpoints of e_i , without loss of generality we assume that q is above or at the same height as p and that p is to the left of q. We distinguish two cases based on σ_i .

First, the case when $\sigma_i \leq 1$, as illustrated in Figure 4a. In this case $|e_i|_x \geq |e_i|/\sqrt{2}$ and $\operatorname{dist}(\mathbf{T}_{i-1}) \geq (|e_i|/\sqrt{2}-\varepsilon)/2$. Filling this into our α -distance property we get

$$1/\alpha \ge \frac{\operatorname{dist}(\mathbf{T}_{i-1})^2}{\operatorname{area}(\mathbf{T}_{i-1})} \ge \frac{(|e_i|/(2\sqrt{2}) - (\varepsilon/2))^2}{|e_{i-1}| \cdot |e_i|} = \frac{|e_i|^2/8 - |e_i| \cdot \varepsilon/(2\sqrt{2}) + \varepsilon^2/4}{|e_{i-1}| \cdot |e_i|}$$
$$\ge \frac{|e_i|^2/8 - |e_i| \cdot \varepsilon/(2\sqrt{2})}{|e_{i-1}| \cdot |e_i|} = \frac{|e_i|/8 - \varepsilon/(2\sqrt{2})}{|e_{i-1}|},$$

which can be rewritten as

$$|e_i| \le 8/\alpha \cdot |e_{i-1}| + 4\varepsilon/\sqrt{2}.$$

In the second case, we have $\sigma_i > 1$, as shown in Figure 4b. Let *s* be a point on the supporting line of e_i that is horizontally aligned to e_0 , then at least one of the edges *ps* or *qs* must have length at least $e_i/2$. If $|qs| \ge |e_i|/2$ we find that $\operatorname{dist}(\mathbf{T}_{i-1}) \ge |e_i|/(2\sqrt{2})$. If this is not the case, then the edge *ps* is a segment of e_i and must be below e_0 and $|ps| \ge |e_i|/2$. Since *ps* cannot intersect the boundary of *P* and its nearest points are on one of the vertical neighbors of e_0 either *p* or *s* has a horizontal distance of at least $|e_i|/(2\sqrt{2}\sigma_i)$ to the boundary. We again fill this into our region property to get

$$1/\alpha \ge \frac{\operatorname{dist}(\mathbf{T}_{i-1})^2}{\operatorname{area}(\mathbf{T}_{i-1})} \ge \frac{(|e_i|/(2\sqrt{2}\sigma_i))^2}{|e_{i-1}| \cdot |e_i|} = \frac{|e_i|^2}{8\sigma_i^2 \cdot |e_{i-1}| \cdot |e_i|} = \frac{|e_i|}{8\sigma_i^2 \cdot |e_{i-1}|}$$

Rewriting this we get

$$|e_i| \le 8/\alpha \cdot \sigma_i^2 |e_{i-1}|.$$

Combining these two we find that

$$\begin{split} e_{i} &| \leq 8/\alpha \cdot (1+\sigma_{i}^{2})|e_{i-1}| + 4\varepsilon/\sqrt{2} \\ &\leq 8/\alpha \cdot (1+((2^{i+1}-2)/\alpha)^{2})|e_{i-1}| + 4\varepsilon/\sqrt{2} \\ &\leq 8/\alpha \cdot (1+2^{2i+2}/\alpha^{2})|e_{i-1}| + 4\varepsilon/\sqrt{2} \\ &\leq 2^{2i+6}/\alpha^{3}|e_{i-1}| + 4\varepsilon/\sqrt{2} \\ &\leq 2^{2i+6}/\alpha^{3} \cdot 8\varepsilon/\alpha^{3(i-1)} \cdot 2^{(i)(i+6)} + 4\varepsilon/\sqrt{2} \\ &= 8\varepsilon/\alpha^{3i} \cdot 2^{(i+1)(i+7)-1} + 4\varepsilon/\sqrt{2} \\ &\leq 8\varepsilon/\alpha^{3i} \cdot 2^{(i+1)(i+7)}. \end{split}$$

Note that in the last step we assume that $\alpha \leq 1$. Which is fine, since higher values of α only make the α -distance property stricter, and we are constructing a lower bound.

Theorem 4 implies that we cannot restrict ourselves to triangulations if we want a linear-size decomposition with the α -distance property. We hence consider possibly non-conforming decompositions (that is, we allow T-junctions) using convex k-gons. We first show how to compute a linear-size decomposition with the α -distance property that uses convex k-gons for $k \leq 7$, and then we argue that each k-gon can be further subdivided into convex quadrilaterals and triangles.

A decomposition with 7-gons. We assume without loss of generality that no two vertices of the input polygon P have the same x- or y-coordinates. We describe a recursive algorithm that computes in each step a single 7-gon³ of the subdivision and then recurses on up to four smaller polygons. In a generic step of the recursive procedure, we are given a polygon bounded by a chain of edges from the original polygon and by two subdivision edges, one vertical and one horizontal; see Figure 5a. (In our figures we use gray lines for subdivision edges, solid black lines for polygon edges, and dotted black lines to indicate an unspecified continuation of the boundary of the input polygon. Black disks mark vertices of the input polygon.) The subdivision edges meet in a vertex, the *corner*

³From now on, when we use the term 7-gon, we mean a convex k-gon for $k \leq 7$.

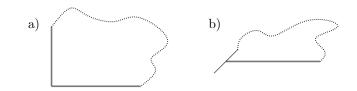


Figure 5: Polygons on which we recurse consist of up to two subdivision edges and a boundary chain.

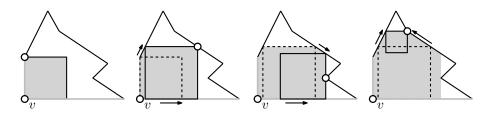


Figure 6: Example of constructing a 7-gon from a recursion polygon.

of the polygon. One of the subdivision edges can have zero length (see Figure 5b). Without loss of generality we assume that the horizontal subdivision edge, $e_{\rm h}$, is the longer of the two subdivision edges, and that the vertical subdivision edge $e_{\rm v}$ extends upward from the left endpoint of $e_{\rm h}$. Initially, P does not have the right form as there are no subdivision edges. Hence we first pick an arbitrary point in the interior of P and shoot axis-aligned rays in all four directions. This subdivides P into four polygons that each have exactly two subdivision edges that meet in a vertex.

We now describe how we generate a 7-gon of the decomposition in a recursive step on input polygon $Q \subset P$ with two subdivision edges, e_h and e_v , meeting in corner v, see Figure 6 for an example. We first grow a square with v as lower-left corner, until the square hits the boundary of Q. (This could be immediately, if the vertical subdivision edge has zero length.) If one of the edges of the square hits a vertex of the original polygon P, we stop. Otherwise a vertex of the square hits an edge of P. We then start pushing the square along the edge, meanwhile growing it so that it remains in contact with the subdivision edge. This again continues until the boundary of P is hit, which may either terminate the process (when a vertex of P is hit) or not (when an edge is hit), and so on. The 7-gon will be the union (swept volume) of all squares generated during the entire process. Figure 7 gives an overview of the cases that can arise, with A being the start configuration. Thick arrows indicate a transition from one case to another. As mentioned, we stop pushing a square when a new vertex of Poccurs on the boundary. Cases where this happens are given a number (A1, B1, $B2, \ldots$). Next we provide more details on how to push the squares in each of the cases and when one case transitions to another. The top left, top right, bottom left, and bottom right vertex of a square will be denoted by $p_{nw}, p_{ne}, p_{sw}, p_{se}$ respectively, and the top, right, bottom, and left edge by $e_{\rm n}, e_{\rm e}, e_{\rm s}, e_{\rm w}$. In each case the process ends when a vertex of P is hit.

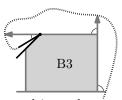
- A We grow a square from the corner while keeping e_s on e_h and e_w on e_v until it hits an edge or vertex of P. We go into case B if p_{nw} hits an edge e_{nw} of P or into case E and F if p_{ne} hits an edge e_{ne} . Note that p_{se} cannot hit an edge of the polygon before p_{nw} , since e_h is at least as long as e_v .
- B The vertex p_{nw} is on an edge e_{nw} of P and e_s is on e_h . We push the square to the right while maintaining these contacts. We go into case C if p_{se} hits an edge e_{se} of P or into case D and F if p_{ne} hits an edge e_{ne} of P.

- C The vertex p_{nw} is on an edge e_{nw} of P and p_{se} is on an edge e_{se} of P. We push the square up and to the right maintaining these contacts. We go into case D and G if p_{ne} hits an edge e_{ne} of P.
- D The vertex p_{nw} is on an edge e_{nw} and p_{ne} is on an edge e_{ne} of P. We push the square upward while maintaining these contacts.
- E The vertex $p_{\rm ne}$ is on an edge $e_{\rm ne}$ of P and $e_{\rm w}$ is on $e_{\rm v}$. We push the square upward while maintaining these contacts. We go into case D if $p_{\rm nw}$ hits an edge of P.
- F The vertex p_{ne} is on an edge e_{ne} of P and e_s is on e_h and we push the square to the right while maintaining these contacts. We go into case G if p_{se} hits an edge of P.
- G The vertex p_{ne} is on an edge e_{ne} and p_{se} is on an edge e_{se} of P, and we push the square to the right while maintaining these contacts.

Lemma 8. The process above generates a convex k-gon C with $k \leq 7$. Moreover, for any $p \in C$ we have $\operatorname{area}(C) \geq \frac{1}{2} \cdot \Delta_p^2$, where Δ_p denotes the distance from p to the boundary of the original polygon P.

Proof. A straightforward case analysis of the different paths that the process may follow in Figure 7—note that we can actually follow several paths, since sometimes we continue pushing in two separate directions—shows that C is a convex 7-gon. The construction guarantees that C is the union of a (possibly infinite) set of squares that each touch the boundary of P. Let p denote a point in C and σ a square containing p that touches the boundary of C. Then $\Delta_p \leq \sqrt{2} \cdot \text{length}(\sigma)$, where $\text{length}(\sigma)$ denotes the edge length of σ . It follows that $\operatorname{area}(C) \geq \operatorname{area}(\sigma) = \operatorname{length}(\sigma)^2 \geq \frac{1}{2} \cdot \Delta_p^2$.

After constructing the 7-gon C, we should recurse on the remaining parts of the polygon. The parts we can recurse on must have at most two orthogonal subdivision edges that meet in a point, as in Figure 5. Parts for which this is not yet the case are first subdivided further by shooting horizontal and/or vertical rays from certain



vertices of C so that the required property holds for the resulting subparts. Which rays to shoot depends on the final case in the construction of C. The figure to the right shows case B3; the corners of the parts on which we recurse are indicated by small circular arcs. In total, we may get up to four parts in which we recursively construct new 7-gons. Next, we bound the total number of regions that are created.

Lemma 9. The algorithm described above creates $O(n_P)$ 7-gons in total, when applied to a polygon P with n_P vertices.

Proof. Let V_Q denote the subset of vertices of P that are on the boundary of a polygon $Q \subset P$ on which we recurse, excluding the possible vertices of P that are the endpoints of the subdivision edges of Q. Recall that after we construct a 7-gon C inside Q, the remainder of Q is subdivided into at most four parts

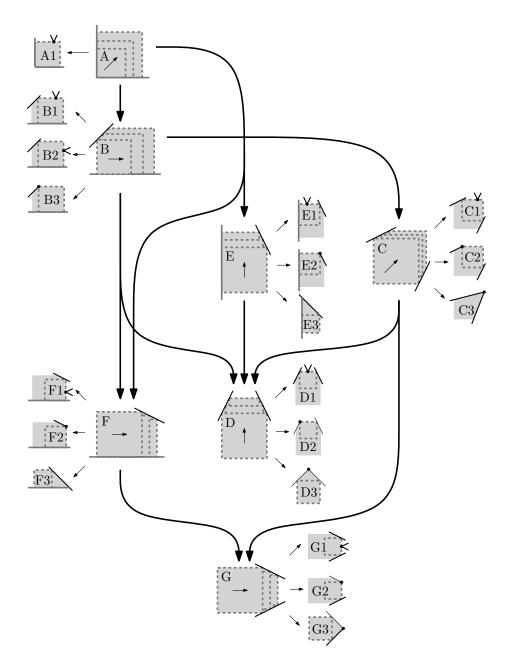


Figure 7: We construct 7-gons by pushing squares through the polygon according to cases A to G. Fat arrows indicate a transition from one case to another and a split means that we continue in two separate directions. Note that cases E and F, and D and G are symmetric.

on which we recurse again. At least one vertex of V_Q is on the boundary of C, so each part has strictly fewer vertices of P on its boundary. We also know that each vertex of V_Q can be on the boundary of at most one part (recall that vertices on endpoints of subdivision edges are not considered). It follows that only $O(n_P)$ 7-gons are constructed.

Next we describe how to implement the algorithm in $O(n_P \log n_P)$ time. Each of the cases A to G can be viewed as moving a square from a start location to an end location such that all intermediate squares have specific contacts to the polygon Q as detailed in each case description. To find the swept volume of this sequence of squares it suffices to know in each of the cases at which squares we start and end. To find these start and end squares we need some supporting data structures.

We use the medial axis \mathcal{M} of P, with the following asymmetric convex distance function. Let p and q be two points in the plane. The distance from pto q is the edge length of the smallest square with its lower left corner on p that has q on its boundary. This is different from the L_{∞} distance, since we grow a square from its corner, not its center. As a result, the "distance" from p to q is defined only if q lies to the north-east of p. However, for any point inside P the distance to the boundary of P and the nearest point on the boundary are well defined, which is sufficient for our purposes. Conceptually, one can also set the undefined distances to infinity.

Such a medial axis is the same as the Voronoi diagram of the line segments that form the polygon boundary with respect to a convex distance function. Fortune showed how to compute a Voronoi diagram of line segments in $O(n \log n)$ time for the Euclidean distance using a sweepline approach [10]. This approach can be extended to convex distance functions, even when the reference point is on the boundary as in our case [9, 11]. We then construct the following data structures:

- We preprocess each medial axis so that we can do point location in $O(\log n)$ time. Since the medial axis is a connected polygonal subdivision this can be done in O(n) time [15].
- We also preprocess each medial axis so that we can answer horizontal and vertical ray shooting queries in $O(\log n)$ time. This can again be done in O(n) time by first computing the horizontal and vertical decomposition of P [6], and then preprocessing these trapezoidal maps for point location.
- Finally, we preprocess P itself in O(n) time such that we can do horizontal and vertical ray shooting in $O(\log n)$ time.

Initially (case A) we want to find the largest square that we can grow from the corner v. We locate the cell of \mathcal{M} that contains v, which gives us the vertex or edge of the polygon, say edge e, that is closest to v in the specified distance measure. This implies that e is the first edge hit by the boundary of a square grown from v. In this way we determine in $O(\log n)$ time if we are in case A, B, or E. Next we push the square upward or to the right. We then have to determine the final square for that movement and in which case we should continue. We distinguish two different types of movement for the square. Either the square has one edge on one of the vertical or horizontal subdivision edges (case B, E, and F), or it has two corners on polygon edges of P (case C, D, and G).

If one edge of the square stays on a subdivision edge then specifically the lower left vertex stays on the subdivision edge and the series of squares that we create are exactly the largest squares with their lower left corners on the subdivision edge. Recall that we stop moving the square when another edge or vertex of P hits the boundary of the square. Let q denote the lower left corner of this square. By definition of \mathcal{M} the point q has to be on a bisector of \mathcal{M} as there are two different features (edges or vertices) of P that are at equal distance. Hence, the process of moving a square along a subdivision edge is essentially the same as moving its lower left corner point until it hits an edge of the medial axis (or P). We can use horizontal or vertical ray shooting to find in $O(\log n)$ time the point q where we end the movement along the subdivision edge.

When we move a square while keeping two vertices on edges of P it follows from the definition of the medial axis and our distance measure that the lower left vertex of the square remains on the bisector of the two edges of P. The movement ends when a third edge or vertex of P is on the boundary of the square, so at a vertex of the medial axis. Specifically the vertex where the bisector along which the lower left vertex was moving, ends. To find the final square of the movement we have to find the bisector, determine which endpoint of the bisector we need and find the three edges or vertices of P that define that vertex. Since we already found the right bisector in the previous case, each of these steps can be done in O(1) time after which we can determine in O(1) time how to continue. To summarize, we obtain the following lemma.

Lemma 10. Computing the 7-gon in a recursive step of the algorithm takes $O(\log n)$ time, after O(n) preprocessing.

From 7-gons to quadrilaterals and triangles. As a last step we can convert the 7-gons from our decomposition into convex quadrilaterals and triangles. The resulting decomposition still has the α -distance property, although the value for α will decrease from 1/2 to 1/8, as shown below. Let Q denote a convex polygon with n_Q vertices. By the ham-sandwich theorem [8], there exists a line cutting Q into two portions of equal area with at most $\lfloor n_Q/2 \rfloor$ vertices of Q strictly on each side of the line. Cutting along this line, we obtain two polygons with half the area and at most $\lfloor n_Q/2 \rfloor + 2$ vertices each. By repeating this process, if necessary, we obtain either triangles or quadrilaterals. Using these ham-sandwich cuts we prove the following theorem.

Theorem 11. Given a simple polygon P we can compute in $O(n \log n)$ time a subdivision of P consisting of O(n) triangles and convex quadrilaterals with the (1/8)-distance property.

Proof. By Lemmas 9 and 10 we can compute in $O(n \log n)$ time a decomposition of P into O(n) convex k-gons, for $k \leq 7$, that has the (1/2)-distance property. We further subdivide each k-gon using ham-sandwich cuts, as explained above.

In the worst case we start with a 7-gon that is split into two 5-gons by the first ham-sandwich cut, after which each 5-gon is split into two quadrilaterals. We then get four quadrilaterals each having 1/4 of the area of the 7-gon. Thus, since the decomposition into 7-gons had the (1/2)-distance property, the new decomposition has the (1/8)-distance property.

Combining this result with Theorem 1 we obtain the following corollary.

Corollary 12. Let S denote a planar polygonal subdivision with O(n) vertices and let γ_i for each $P_i \in S$ denote the probability that a query point lies in P_i . We can construct in $O(n \log n)$ expected time a point location structure that uses O(n)space and answers a query with a point p in $O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$ time, where Δ_p denotes the Euclidean distance from p to the nearest point on any edge of S.

3. Depth-bounded quadtree

Although computing a distance-sensitive decomposition takes $O(n \log n)$ time asymptotically there is a lot of overhead involved. During preprocessing we need several medial axes of the input subdivision S, and each of these has to be further processed for point location and horizontal and vertical ray-shooting. We also create many additional regions which would cause the worst-case $O(\log n)$ search time to have a much larger constant when compared to a worst-case optimal point location structure. In this section we present a much simpler solution that has very little extra overhead compared to a general worst-case optimal point locations structure, but only works for a special case of the problem.

In this special case we assume no distribution of the queries over the polygons of the subdivision is given and we want the query time to be dependant only on the distance from a point to the boundary. Let S be a planar polygonal subdivision and assume that S is contained in a square with area 1. (Note that in this case we do not require S to be connected.) We show how to construct a query structure that can answer a query for a point p in $O\left(\min\left(\log n, 1 + \log \frac{1}{\Delta_p^2}\right)\right)$ time, where Δ_p again denotes the Euclidean distance from p to the nearest point on any edge of S. This can be seen as a special case of the general problem where each region $P_i \in S$ has a weight proportional to its area, so $\gamma_i = \operatorname{area}(P_i)/\operatorname{area}(S)$.

In essence we have two different requirements for a query. First, no query should ever take more than $O(\log n)$ time, and second, a query for a point far from the boundary should take only $O(1 + \log \frac{1}{\Delta_p^2})$ time. A worst-case optimal point location structure can be used to satisfy the first requirement and a quadtree where each leaf intersects O(1) features of the subdivision satisfies the second requirement. Unfortunately, neither satisfies both: a quadtree may have nodes with a very high depth and a worst-case optimal point location structure gives no guarantees on finding points far from the boundary quickly. We can however use both structures together to get the bound we need.

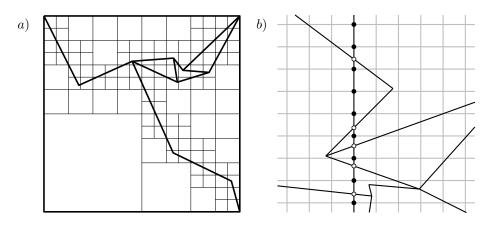


Figure 8: a) A depth-bounded quadtree $\mathcal{QT}(S)$. b) An illustration of the sweep-line algorithm. Closed disks indicate grid vertices and open disks indicate intersection points of subdivision edges with the sweep-line.

We construct two structures: a general worst-case optimal point-location structure $\mathcal{PL}(S)$ and a depth-bounded quadtree $\mathcal{QT}(S)$. With a slight abuse of terminology we use *leaf, root* and *node* to denote nodes of the quadtree as well as the square regions they are associated with. The root of the quadtree is the bounding square of S, which we assume to have edge length 1. Each leaf of the quadtree is either empty—it does not intersect the boundary of S—or it has a depth of $\lceil \log \sqrt{n} \rceil$; see Figure 8a. A query for a point p first finds the leaf v of the quadtree that contains p. If v does not intersect any of the boundary elements of S, then the polygon $P \in S$ that contains v also contains p. If v is not empty, then we conclude that p is close to the boundary of S and perform a query in $\mathcal{PL}(S)$.

Preprocessing. Constructing a worst-case optimal point-location structure takes $O(n \log n)$ time, where n is the complexity of S. When constructing the quadtree we have to account for the presence of edges of S, and not just its vertices. The standard method to construct a quadtree on a set of points is to recursively split nodes that contain more than one point and propagate the points down the tree such that each leaf stores the points contained in its associated square. In our case each leaf would have to store the edges that intersect it, which would lead to superlinear storage as each edge may intersect many leaves of the quadtree. Instead we use a different approach that uses a sweep-line over the underlying grid of the quadtree.

We first construct the complete quadtree up to depth $\lceil \log \sqrt{n} \rceil$, which represents a grid where each cell has an edge length ℓ between $1/(2\sqrt{n})$ and $1/\sqrt{n}$. It follows that the grid contains O(n) cells in total. We will mark each leaf of the quadtree whose associated grid-cell is intersected by an edge of S. A cell of the grid is intersected by an edge of S if and only if either one of its boundary segments intersects an edge or if the cell contains a vertex of S. We can mark leaves that contain a vertex by locating each vertex within the grid, which takes $O(n \log n)$ time. We then use two sweep-lines to mark cells whose boundary segments are intersected by edges of \mathcal{S} . We use a horizontal sweep-line to mark all leaves whose grid cells have their left or right boundary segment intersected by an edge of \mathcal{S} . The sweep goes from left to right and we maintain an ordered list of edges from the subdivision that intersect the sweep-line. This ordering changes only when the sweep-line encounters vertices of the subdivision. When the sweep-line encounters a vertex v we locate the vertex in the current edge-ordering in $O(\log n)$ time and then spend $O(k \log n)$ time adding and removing edges adjacent to v, where k is the degree of v. As there are O(n)vertices and the sum of their degrees is O(n) the vertex events take $O(n \log n)$ time in total. When the sweep-line encounters a vertical line of the grid we test for intersections between the subdivision edges stored in the sweep line and the vertical grid-segments that coincide with the sweep line. Each vertical grid-segment—the boundary edge of one or two cells—is intersected if and only if there are edges of \mathcal{S} between its endpoints on the sweep-line. This is easy to test by simply locating each grid-vertex on the vertical line in the edge-ordering stored in the sweep-line; see Figure 8b. If a grid-segment is intersected by an edge of the subdivision we mark the leaves whose cells are to the left and right of this grid-segment. For each such event we have to perform $O(\sqrt{n})$ binary searches on the edge-ordering of the sweep-line, taking $O(\sqrt{n} \log n)$ time in total. Since there are $O(\sqrt{n})$ such events this takes $O(n \log n)$ time in total. This sweep marks all cells of which the left or right boundary edge is intersected by a subdivision edge. A similar vertical sweep is used to mark all leaves of which the top or bottom segment of its associated grid-cell is intersected by a subdivision edge.

After performing both sweeps each leaf intersected by the subdivision boundary is marked. Next we mark internal nodes of the quadtree of which the associated square intersects an edge of S. We use a bottom-up approach where each node is marked if and only if at least one of its children is marked. Next, the tree is trimmed by removing all nodes with an unmarked parent. The resulting quadtree is a depth-bounded tree in which each leaf has depth $\lceil \log \sqrt{n} \rceil$ or does not intersect the boundary of the subdivision S. As a final step we do a single point location for each empty (not marked) leaf of the quadtree to determine which polygon of S it is contained in and store this information in the leaf.

Lemma 13. Given a subdivision S, we can construct the depth-bounded quadtree QT(S) and worst-case optimal point-location structure PL(S) in $O(n \log n)$ time, where n is the complexity of S.

Querying. Given the quadtree $\mathcal{QT}(S)$ and the point location structure $\mathcal{PL}(S)$ we perform a point location query on a point p as follows. We first find the leaf v of $\mathcal{QT}(S)$ that contains p. If v is empty, then we report the polygon that contains v, otherwise we do a point location query for p in $\mathcal{PL}(S)$ to find the polygon containing p. Next we show that this indeed provides us with the required query-time.

Lemma 14. A point-location query as described above for a point p takes $O(\min(1 + \log \frac{1}{\Delta_p^2}, \log n))$ time, where Δ_p is the distance from p to the boundary of S.

Proof. We distinguish two cases. First assume the leaf v from $\mathcal{QT}(S)$ that contains p is empty. Let i denote the depth of v in the quadtree, so we spend O(i) time to locate p. The node v has an edge length of $1/2^i$, and its parent has an edge length of $2/2^i$. The parent of v was split, so it must have intersected the boundary of S. This implies that $\Delta_p \leq 2\sqrt{2}/2^i$, since both p and some point on the boundary of S are contained in the parent of v. Plugging this in, we find that indeed

$$O(i) = O\left(\min\left(1 + \log\frac{1}{2\sqrt{2}/2^i}, \log n\right)\right) = O\left(\min\left(1 + \log\frac{1}{\Delta_p^2}, \log n\right)\right).$$

Now suppose v is not empty. In this case we spend $O(\log n)$ time in the quadtree and $O(\log n)$ time in the general point location structure. However, since v must have an edge length of at most $1/\sqrt{n}$ and is intersected by the boundary of S we know that $\Delta_p \leq \sqrt{2}/\sqrt{n}$ and the query bound follows. \Box

Combining Lemmas 13 and 14, we obtain the desired result.

Theorem 15. Given a planar piecewise-linear subdivision S contained in a square with edge length 1, we can construct in $O(n \log n)$ expected time a point location structure that can answer a query for a point p in S in $O(\min(1 + \log \frac{1}{\Delta_p^2}, \log n))$ time, where Δ_p denotes the distance from p to the boundary of the polygon $P \in S$ that contains it.

Convex subdivisions in \mathbb{R}^3 . The above method of using a depth-bounded quadtree together with a worst-case optimal point-location structure can also be applied to convex subdivisions in \mathbb{R}^3 . In this case we would want to compute a depthbounded octree, where each leaf either does not intersect any boundary facet or has depth $\left[\log \sqrt[3]{n}\right]$. As before we can first construct the full octree of depth $\left[\log \sqrt[3]{n}\right]$ and then mark leaves that intersect the subdivision boundary. In a general connected subdivision in 3D a cell is intersected if and only if its 2-dimensional faces are intersected by a subdivision facet. The straightforward extension of the sweep-line approach from the 2-dimensional case would require us to maintain a dynamic subdivision defined by the intersection of the input subdivision \mathcal{S} and the sweep-plane. Then whenever the sweep-plane encounters a plane in the grid we should determine if the boundary squares of the grids cells are empty in the sweep-plane. This seems difficult to do in near-linear time, as we cannot afford to traverse the entire sweep-plane, which may have $\Theta(n)$ complexity. However, in a convex subdivision a grid cell is intersected by a subdivision facet if and only if at least two of its vertices are in different cells of the subdivision. As a result we can simply perform a point location query on each vertex of the grid and test for each grid cell whether all vertices are contained in the same polyhedron of the subdivision. If not all vertices belong

to the same polyhedron we mark the associated leaf of the octree. We can use the $O(n \log n)$ space structure by Snoeyink [22] to perform each query in $O(\log^2 n)$ time. After marking the leaves of the octree we propagate the marking upwards, trim the tree and determine for each empty leaf which polyhedron contains it, similar to the two-dimensional case. A query for a point p is again performed by first locating p in the octree, where at most $O(\log n)$ time is spent. If the resulting leaf is not empty we instead find p in the general point location structure in $O(\log^2 n)$ time.

Theorem 16. Given a 3-dimensional convex polyhedral subdivision S contained in a cube with edge length 1, we can construct in $O(n \log^2 n)$ time and $O(n \log n)$ space a point location structure that can answer a query for a point p in Sin $O(\log \frac{1}{\Delta_p^2})$ time if $\Delta_p \geq \sqrt{3}/\sqrt[3]{n}$ and $O(\log^2 n)$ otherwise, where Δ_p is the shortest distance from p to nearest boundary facet of S.

4. Conclusions

We presented two data structures for distance-sensitive point location. The first and most general structure relies on decomposing a connected planar subdivision into constant complexity regions, such that any point that is far from the boundary is contained in a large region. We then showed how such a distance-sensitive decomposition is used to create a distance-sensitive point location structure. Computing the decomposition and the point location structure takes $O(n \log n)$ time and O(n) space. A query for a point p with distance Δ_p to the nearest point on an edge of S takes $O\left(\min\left(\log n, 1 + \log \frac{\operatorname{area}(P_i)}{\gamma_i \Delta_p^2}\right)\right)$ time, where γ_i denotes the given probability that a query falls in polygon P_i . Our distance-sensitive decomposition consists of triangles and quadrilaterals and may be non-conforming, that is, there may be T-junctions along their boundaries. An obvious questions is whether "nicer" decompositions are possible that have the same property that a point far from the boundary is guaranteed to be in a large region. We showed that if we insist on a conformal Steiner triangulation, then we cannot bound the number of regions as a function of n, the number of edges of \mathcal{S} . For non-conformal triangulations or conformal quadrilaterals this questions is still open. Another interesting open question is if a similar decomposition is possible for subdivisions in three dimensions. Note, however, that this would not directly lead to distance-sensitive point location structure since, to our knowledge, no three-dimensional entropy-based point location structures are known.

We also presented a simpler structure that does not take into account the query distribution between different regions of the input subdivision S. Instead only the distance from a query point to the nearest edges of the subdivision is considered. This can be seen as a special case of the general distance-sensitive problem, where each each polygon P_i has a probability $\gamma_i = \operatorname{area}(P_i)$. The point-location structure consists of a quadtree with a maximum depth of $\lceil \log \sqrt{n} \rceil$ and a general worst-case optimal point location structure, both of which can

be constructed in $O(n \log n)$ time and O(n) space. A query for a point p then takes $O(\min(\log n, 1 + \log \frac{1}{\Delta_p^2}))$ time. This is not asymptotically better than if we would use the general solution, but we believe this second structure is much simpler to construct and has a smaller overhead. The quadtree-based structure can also be extended to work for convex subdivisions in three dimensions. It takes $O(n \log^2 n)$ time and $O(n \log n)$ space to construct a worst-case efficient structure and a depth-bounded octree. A query then takes $O(1 + \log \frac{1}{\Delta_p^2})$ time if $\Delta_p \geq \sqrt{3}/\sqrt[3]{n}$ and $O(\log^2 n)$ time otherwise. Note that the $O(n \log n)$ space requirement comes from the worst-case efficient point location structure as no O(n) space structure is yet known that has $O(\log^2 n)$ query time.

5. References

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