



Brock University

Department of Computer Science

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Vlad Wojcik and Pascal Comte  
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Brock University  
Department of Computer Science  
St. Catharines, Ontario  
Canada L2S 3A1  
[www.cosc.brocku.ca](http://www.cosc.brocku.ca)

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## **ABSTRACT:**

*The real-time tasks of image interpretation, pattern clustering, recognition and classification necessitate quick analysis of registered patterns. Current algorithms and technological approaches are slow and rigid. Within the evolutionary context we present here a novel, biologically inspired solution to these problems, with aim to accelerate processing. Our massively parallel algorithms are applicable to patterns recorded on rectangular imaging arrays of arbitrary sizes and resolutions. The systems we propose are capable of continuous learning, as well as of selective forgetting of less important facts, and so to maintain their adaptation to an environment evolving sufficiently slowly. A case study of visual signature recognition illustrates the presented ideas and concepts.*

## **1. The challenge**

Real-time robotic understanding of perceived images is exceedingly challenging and demands high computing power and novel, parallel algorithms. The processing power available today is woefully scarce. As a case in point consider the speed of leading professional cameras offered currently by market leaders: Nikon and Canon. Compare the film camera model Nikon F6, capable of shooting 8 frames per second, to its digital counterpart, model D3, which offers the resolution of 12 megapixels, and shoots at 9 fps rate [1]. (Faster shooting rates are available in the resolution crop mode) Interestingly, if we scan a single full frame of the old 35 mm film with Microtek ArtixScan M1 scanner at 4800 ppi resolution we can obtain a 36 megapixel image [2].

Using that scanner a 6x7 cm medium format film frame yields 130 megapixel scan of astounding quality, which, when saved, produces a 400 MB TIFF file. Note that this resolution is roughly equivalent to that of the human retina. Scanners offering higher resolutions exist. The challenge is to process such images in real time, to offer digital IMAX 3D cinema first, and real-time adaptive 3D robotic vision later.

Canon cameras fare no better: The top film model EOS-1v is capable of shooting at 10 fps, while its digital counterpart EOS-1Ds Mark III (a dual-processor, 21 megapixel camera) features the shooting rate of up to 5 fps [3].

Normally we expect mechanical devices to be slower when compared to electronic devices, which do not have moving parts. Yet in these examples, the mechanical still film cameras are typically faster than their digital counterparts, precisely because digital cameras need extra time to process, compress and store the images they record, although no attempt is made to understand the images. That would not be practical, given current state of technology.

It is worth mentioning here that some cameras feature a simple face recognition technology in their focus modes: without attempting to identify individuals the camera strives to keep in focus as many human faces as possible.

High definition video (HDTV) does not fare any better. While the frame rate is higher (24 fps), the image resolution is substantially lower: 1080x1920 pixels, i.e. about 2 megapixels per frame. Video cameras of higher resolution exist [4], but there is no commercial way to broadcast such signals.

Further improvements in image processing intelligence and operating speed call for novel algorithmic approaches, stemming from deeper understanding of biological vision.

We are witnessing the trend of increasing the resolution of digital imaging devices. However, doubling the camera linear resolution leads to quadrupling of the number of pixels of the imaging sensor. To keep the image processing times constant, we must significantly increase cameras' computing power needed to interpret, compress and store the images.

Consider any two arbitrary patterns made of  $N$  black pixels on a white background. Suppose we are to decide whether the patterns are sufficiently similar to be grouped in the same category. For instance: given two signatures we need to decide whether they belong to the same person. We cannot *a priori* rely on any signature feature like alphabet, penmanship, signature owner's literacy level, etc. In this comparison every pixel may count: We have no choice but to compare every pixel of one signature against every pixel of another signature. This calls for an algorithmic performance of  $O(N^2)$ , where  $N$  is the number of pixels on camera's imaging sensor. If we double the linear resolution of the imagers, the number of pixels in each signature would quadruple  $N$  and we could expect the time to classify the signatures to increase sixteen fold. We need all computing power we can get! After all, with increasing image resolution we can render finer details and so increase the reliability of our classification process.

However, the chip technology seems to have reached a plateau in terms of computing power it can extract from a single on-chip CPU: By now it is widely accepted that we can no longer expect to double computing speed every 18 months or so.

We have no choice but to rely on parallel computing if we ever want to have robots capable of real-time image understanding.

This technological snapshot begs for a biological comparison. Consider human visual performance: each of our retinas has approx. 125 million sensors, and with binocular vision we are therefore capable of real-time processing of the continuous stream of 250 megapixel information. Although visually oriented, humans are far from the pinnacle of visual acuity. Unlike us, both birds and insects can see in four basic colours – including ultraviolet [5]. Even an octopus has more sensors on its retinæ than a human – although it is shortsighted (there is no evolutionary advantage of long-sight in water). A mantis shrimp is equipped with hyperspectral vision: it has 16 different

photoreceptor types, capable of distinguishing between 12 bandwidths of light (including IR and UV) and of various polarizations [6]. It has a tiny brain, and an even smaller “visual cortex”. To survive it must run simple, fast and efficient algorithms to handle the information it receives.

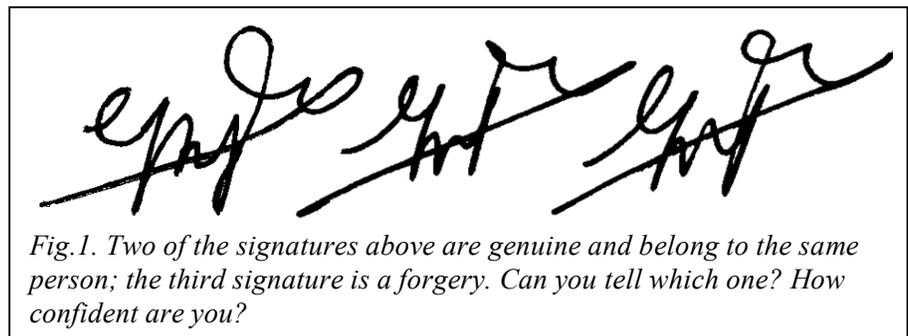
We (the humans) have read many sophisticated papers on computer vision, full of intricate mathematics. While the authors of this paper fully admire the intellectual agility and mathematical acumen of our colleagues working on vision, we would like to argue that visual information must be relatively easy to process, if it is to be of use to mantis shrimps, snails, insects, etc. Such relatively simple animals do not have the computing power needed to run complex algorithms proposed so far by humans. The authors of this paper would like to suggest a simpler approach, using only set theory and the concept of metric spaces.

To make our technology mimic the animal performance is a tall challenge. Let’s draw our inspirations from biology:

Animal vision systems are different than current computer vision systems: the visual cortex has many more processing elements (neurons) than sensors (rods and cones) on the retina. The retinal sensors are arranged on a hexagonal grid – this facilitates image understanding. The sensors on artificial retinas (CCD, CMOS devices, etc.) lie on a rectangular grid.

Finally, our (animal) image understanding algorithms do not appear to allocate equal level of attention (is that computing power?) to all elements of an image. To illustrate this, consider the following problem, depicted on Fig.1. Given three signatures, two of them genuine and the third being a forgery, can you tell which one is a fake?

Observe how you work: Having determined that the signatures are penned in black on white background, you pay no attention at all to white pixels. Moreover, you give similar fragments of the signatures less of your attention; you focus on the dissimilarities.



Intuitively: if the dissimilarities between two signatures exceed our common sense “level of tolerance” we conclude that the same person did not pen them. Given the information provided, we group two most similar signatures and label them genuine leaving the remaining one to be a forgery.

Note that if the signatures were in white on a black background, we would not pay attention to black. Actually, a brief attention is needed to distinguish between the foreground and the background of these images. The color we choose as foreground contains fewer pixels. Thereafter, we tend to focus our attention on small sub-groups of pixels bearing largest dissimilarities. This limits our attention effort (computing power?) needed to distinguish between signatures.

In this paper we present computing algorithms exhibiting similar emergent behaviour: most computing power is allocated to small but significant regions of images being compared. We will demonstrate our theory by comparing results of signature verifications. We use signatures merely as

an idealized but practical case study: signatures form images that are small, easily obtainable but unpredictable. One cannot assume anything about the appearance of a signature of a particular person: we cannot presume the use of any particular alphabet, writer's literacy, writing direction, sequence and dynamics, signer's biometrics, etc., so our signature authentication algorithms must be sufficiently general.

We chose to demonstrate our vision methodology using signatures because we can (initially) avoid the non-trivial task of distinguishing between foreground and the background, as well as the problems of identification of objects, perhaps rendered in colour or shades of gray, perhaps in full view or partially occluded, perhaps cast against an unknown background, etc. We know how to address these problems, but we leave this for another paper.

If you, our reader, are still interested to know which signature was a counterfeit: It was the first one.

## 2. A quick review of metric spaces

Intuitively, a metric space is a set of objects, usually called "points", between which a way of measuring the distance has been defined. In everyday life we use Euclidean distance. However, this is only one of the many possible ways of measuring distances.

Let  $\mathbf{S}$  be our space of interest.

**Def. 1. Measure of distance:** Any real-valued function  $d : \mathbf{S} \times \mathbf{S} \rightarrow \mathbf{R}$  can be used as a measure of distance, provided that for all  $x, y, z \in \mathbf{S}$  it has the following properties:

- (2.1)  $d(x, x) = 0$
- (2.2) if  $x \neq y$  then  $d(x, y) > 0$ , i.e. the distance between two distinct points cannot be zero;
- (2.3)  $d(x, y) = d(y, x)$ , i.e. the distance does not depend on the direction of measurement;
- (2.4)  $d(x, z) \leq d(x, y) + d(y, z)$ , i.e. the distance cannot be diminished by measuring it via some intermediate point.

The last property is frequently called the *triangle law*, because the length of each side of a triangle cannot be greater than the sum of lengths of two other sides.

Observe further that properties (2.1) and (2.2) imply that the distance between any two points cannot be negative, and is actually positive if the points are different.

Example: Consider a two-dimensional space  $\mathbf{R}^2$  and two points  $a = \langle x_1, y_1 \rangle$  and  $b = \langle x_2, y_2 \rangle$ .

We may define distance  $d(a, b)$  as:

- (2.5) Euclidean distance:  $d_1(a, b) = \text{sqrt}((x_1 - x_2)^2 + (y_1 - y_2)^2)$  or
- (2.6) Manhattan distance:  $d_2(a, b) = |x_1 - x_2| + |y_1 - y_2|$  or
- (2.7) perhaps simply as  $d_3(a, b) = \max\{|x_1 - x_2|, |y_1 - y_2|\}$

All three functions  $d_1$ ,  $d_2$ , and  $d_3$  meet the necessary requirements (2.1), (2.2), (2.3) and (2.4).

**Def. 2.** A **metric space** is the configuration  $\langle \mathbf{S}, d \rangle$ , where  $\mathbf{S}$  is a set of points, and  $d$  is some measure of distance between them.

In our study we will be frequently interested in measuring distance between subsets of  $\mathbf{S}$ , rather than merely between points of  $\mathbf{S}$ . One of the simplest subsets of  $\mathbf{S}$  is a *ball*.

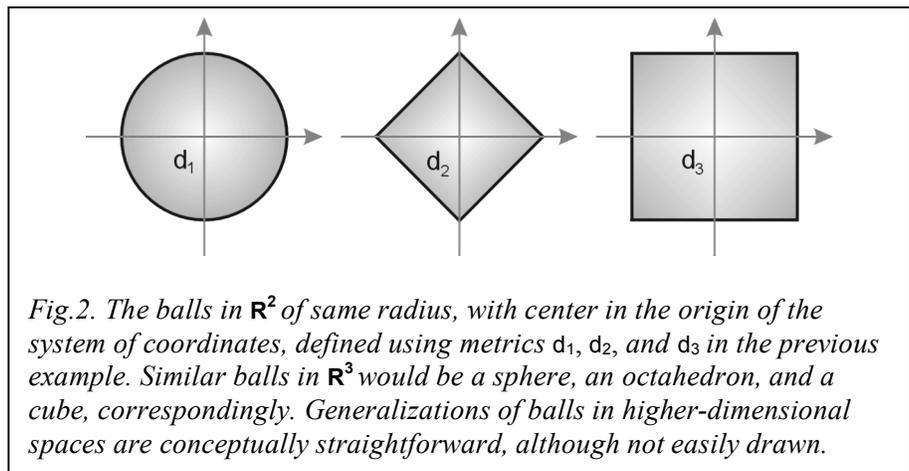
**Def. 3.** A **ball** of radius  $r \geq 0$  around the point  $c \in \mathbf{S}$  is the set

$$(2.8) \quad \{ x \in \mathbf{S} \mid d(c, x) \leq r \}$$

which we will denote as  $B(c, r)$  without mentioning either  $\mathbf{S}$  or  $d$  when confusion can be avoided. The point  $c$  is called the center of the ball  $B$ .

Note: In older math textbooks the term “sphere” is frequently used instead of a “ball”. This usage is currently being phased out, as we prefer now the “sphere” to mean the surface of a “ball”.

Observe that the “shape” of a ball depends on the way we measure distance, as per Fig.2.



It is worth noticing that the “volumes” of such balls may depend on the metrics used. In particular, the Manhattan ball (b) is most specific about its center, i.e. it has the smallest “volume” (i.e. area in  $\mathbf{R}^2$ ), and also its metric function computes faster than metrics (a) and (b). This is important, given that our pattern recognition algorithms, although fast and massively parallel, will remain computationally intensive.

A more detailed analysis of metric spaces can be found at [7].

### 3. On computing distances between sets

Given any two non-empty sets  $A, B \subset \mathbf{S}$  we need to construct a function  $D(A, B)$  to measure distance between  $A$  and  $B$ . That function should retain the properties (2.1), (1.2), (2.3) and (2.4). In particular, observe that the properties (2.1) and (2.2) imply that  $D(A, B) > 0$  even if the sets  $A, B$  touch (i.e. have one common element), or perhaps one of them contains another ( $A \subset B$  or  $B \subset A$ ). In fact, we must construct function  $D$  such that  $D(A, B) = 0$  if and only if  $A = B$ .

Let  $d$  be our chosen function for measuring distance between points of space  $\mathbf{S}$ . We will use that function to construct our function  $D$ .

**Def.4. Distance between a point and a set:** Let  $\langle \mathbf{S}, d \rangle$  be a metric space and let  $A \subset \mathbf{S}$  be a non-empty set. A distance between a point  $x \in \mathbf{S}$  and  $A$ , denoted  $\delta(x, A)$  is given by

$$(3.1) \quad \delta(x, A) = \inf \{ d(x, a) \mid a \in A \}$$

It is the distance between  $x$  and a point  $a \in A$  closest to  $x$ . Using  $\delta$ , let us define

**Def.5. Pseudo-distance between two sets:** Let  $\langle \mathbf{S}, d \rangle$  be a metric space and let  $A, B \subset \mathbf{S}$  be two non-empty sets. A pseudo-distance from  $A$  and  $B$ , denoted  $\Delta(A, B)$  is given by

$$(3.2) \quad \Delta(A, B) = \sup \{ \delta(a, B) \mid a \in A \}$$

In other words, pseudo-distance from  $A$  and  $B$  is the distance from the most distant point  $a \in A$  to  $B$ . It is not distance, but merely pseudo-distance, because it is uni-directional, i.e. the property (2.3) does not hold, given that  $\Delta(A, B) \neq \Delta(B, A)$  in general. To make it bi-directional, we introduce:

**Def.6. Distance between two sets:** Let  $\langle \mathbf{S}, d \rangle$  be a metric space and let  $A, B \subset \mathbf{S}$  be two non-empty sets. A distance between  $A$  and  $B$ , denoted  $D(A, B)$  is given by

$$(3.3) \quad D(A, B) = \max \{ \Delta(A, B), \Delta(B, A) \} \text{ or, perhaps,}$$

$$(3.4) \quad D(A, B) = \Delta(A, B) + \Delta(B, A)$$

Observe that with the information regarding the spatial positioning of sets deliberately destroyed through preprocessing (i.e. using linear transformations), the function  $D$  can be seen as measure of dissimilarity between sets.  $D(A, B) = 0$  implies  $A = B$ . The function  $D$  can be also used as a pattern classifier. With properly selected small value of  $\varepsilon$ , the condition  $D(A, B) \leq \varepsilon$  implies that  $A$  and  $B$  are sufficiently similar to be included in the same category.

## 4. On computing dissimilarities between patterns

We started with an intuitive illustration: Consider two signatures  $A$  and  $B$ , one of them being suspicious. Are they penned by the same person? To determine this you might photocopy each signature on a separate sheet of transparent film. Looking at both films against light, align them for signatures to coincide as much as possible, and decide whether they are similar enough. If so (i.e.  $D(A, B) \leq \varepsilon$ ), you conclude that they were made by the same person. Otherwise you detect a mismatch.

When comparing patterns for the purposes of clustering, recognition and classification, in this paper we will always assume that these patterns are non-empty and have been preprocessed (by shifts and, perhaps, rotations and scalings) so that they coincide as much as possible. Having ensured this we can treat the value  $D(A, B)$  as a measure of dissimilarity between two patterns.

In fact, whether rotations are allowed or not is application dependent. For example: we are unable to recognize our relatives in a photograph presented to us upside down. It seems that our brains do not readily perform rotations of image patterns. However, when trying to recognize a coin we may

rotate it in our hands to obtain a “better view”. Similar argument may be presented on zooming in or out of images (i.e. scaling).

To ensure computability we will assume that the space  $\mathbf{S}$  is discrete. Furthermore, in keeping with the technology of the day we will assume that the points of  $\mathbf{S}$ , which we will call pixels, are arranged in infinite rows and columns. (This constitutes a deviation from the biological systems, where “pixels” are arranged on a hexagonal grid). We will call  $\mathbf{S}$  the “pixel plane”.

Our patterns consist of finite sets of pixels on the pixel plane  $\mathbf{S}$ . Each pixel  $p \in \mathbf{S}$  has a number of attributes: its X and Y coordinates, which we denote  $p.x$  and  $p.y$ , and its exposure value, determining whether or not it belongs to a given pattern.

For the time being, we are considering binarized patterns only, i.e. images of purely black and white (no shades of gray allowed). For this purpose we consider the exposure value to be Boolean, with TRUE indicating that a pixel belongs to a pattern, and FALSE denoting a background pixel. Later we may consider images with levels of gray. In that case the exposure value will be an integer in some range (typically 0 .. 255). For colour images a vector of three such values will be needed, indicating red, green and blue exposure levels. Whatever the type of the pixel value, its value is returned by the function  $val(p)$ .

Our intuitive approach may now be more precise, but still, for two patterns  $A, B \subset \mathbf{S}$ , each consisting of  $N$  pixels, the computation of  $D(A, B)$  is still  $O(N^2)$ . It is time to accelerate computations. We will do it by removing pixels common to  $A$  and  $B$ .

**Theorem 1.** Given two non-empty, pre-processed patterns  $A$  and  $B$ ,  
if  $A \setminus B = \emptyset$  then  $D(A, B) = \Delta(B, A)$

**Proof:** From the definition (3.2) we have  $\Delta(A, B) = \sup \{ \delta(a, B) \mid a \in A \}$ , where  $\delta(a, B) = \inf \{ d(a, b) \mid b \in B \}$ . But for all  $a \in A$  we also have  $a \in B$  (i.e. while we are in  $A$  we are already in  $B$ ), so therefore  $\delta(a, B) = 0$  thus  $\Delta(A, B) = 0$ . Given that  $D(A, B) = \max \{ \Delta(A, B), \Delta(B, A) \}$ , as per (3.3), we have  $D(A, B) = \max \{ 0, \Delta(B, A) \} = \Delta(B, A)$ . ■

Let us establish as well:

**Lemma 1:** Given two non-empty, pre-processed patterns  $A$  and  $B$ : if  $A \setminus B = \emptyset$  and  $B \setminus A = \emptyset$  then  $D(A, B) = 0$ .

**Proof:** Observe that  $A = B$  iff  $A \setminus B = \emptyset$  and  $B \setminus A = \emptyset$ . Therefore we have  $D(A, B) = 0$  from the definition of  $D(A, B)$  (see (3.3)) or, alternatively, by applying the preceding theorem twice, in the direction from  $A$  to  $B$ , as well as from  $B$  to  $A$ . ■

We will follow this intuition and keep removing common pixels. The more similar two patterns are, the more common pixels we will be able to remove. If patterns are woefully dissimilar, they will have relatively few common pixels. These facts point to the potential viability of statistical strategy in pattern matching.

In this paper we will focus on the exact methodology.

**Theorem 2.** Consider two non-empty, pre-processed patterns A and B.  
If  $A \setminus B \neq \emptyset$  and  $B \setminus A \neq \emptyset$  then  $0 < D(A, B) \leq D(A \setminus B, B \setminus A)$ .

**Proof:** From  $A \setminus B \neq \emptyset$  it follows that there exist  $a \in A$  such that  $a \notin B$ . The path from those  $a \in A$  to their nearest  $b \in B$  must be of non-zero length, therefore  $D(A, B) > 0$ .

Consider an arbitrary pixel  $a \in A$ . Its distance to B is  $\delta(a, B) = \inf \{ d(a, b) \mid b \in B \}$ . It may belong to B as well, in which case its distance to B is zero. We ignore such pixels.

If our arbitrary pixel  $a \in A$  does not belong to B, then having removed shared pixels as possible destination points in B, we can only increase the distance from a to  $B \setminus A$ , therefore  $\delta(a, B) \leq \delta(a, B \setminus A)$ .

Identical logic applies in the direction from B to A, yielding  $\delta(b, A) \leq \delta(b, A \setminus B)$ .

Putting all findings together we have:

$$\begin{aligned} 0 < D(A, B) &= \max \{ \Delta(A, B), \Delta(B, A) \} = \max \{ \sup \{ \delta(a, B) \mid a \in A \}, \sup \{ \delta(b, A) \mid b \in B \} \} \leq \\ &\leq \max \{ \sup \{ \delta(a, B \setminus A) \mid a \in A \setminus B \}, \sup \{ \delta(b, A \setminus B) \mid b \in B \setminus A \} \} = \\ &= \max \{ \Delta(A \setminus B, B \setminus A), \Delta(B \setminus A, A \setminus B) \} = D(A \setminus B, B \setminus A), \text{ thus proving that} \end{aligned}$$

$$0 < D(A, B) \leq D(A \setminus B, B \setminus A) \quad \blacksquare$$

Consequently, when comparing two patterns A and B for similarity, instead of calculating  $D(A, B)$  we will remove common pixels first and compute  $D(A \setminus B, B \setminus A)$ . For any  $\varepsilon$ , the result  $D(A \setminus B, B \setminus A) \leq \varepsilon$  guarantees  $D(A, B) \leq \varepsilon$ . With similar patterns rendered in high resolution many common pixels can be removed. A removal of N such pixels accelerates computations by the order of  $O(N^2)$ .

Granted, the removal of shared pixels is not without computational cost. We will demonstrate later in this paper that careful choice of data structures allows to limit this cost to  $O(N)$ .

Further savings in computational complexity and time are still possible by careful selection of the metric function d. Our goal is to remove even more pixels, belonging to those parts of patterns that contain dissimilarities (refer to Fig.1 again).

By now our alert reader has probably figured out where we are going with this reasoning: Given that the problem of pattern comparison is of  $O(N^2)$  nature, by reducing N linearly we reduce its computational cost in proportion to  $N^2$ . The data that will remain we plan to process in a massively parallel fashion. Our approach aims both at speeding up the computations and reducing the size of the “brain” performing them.

## 5. Some useful properties of Manhattan metric

Our preceding reasoning was independent of any metric of distance between points – (2.5), (2.6) or (2.7) were just examples. For practical reasons we pick Manhattan metric (2.6) as our working tool. Our rationale: the code of this metric executes fastest on a digital computer, compared to other examples, and this metric is most selective – it generates the balls of smallest “areas” (see Fig.2). More formally, the ball described by the metric (2.6) in Fig.2 can be inscribed within other balls



from  $a$  to  $b$  via  $c$  for which the equality (5.3) holds. In fact, there are many such trajectories, all of them “straight”!

The only time we will have no choice but to pay the distance penalty (5.4) on our travels from  $a$  to  $b$  is when choosing to stop at  $c$  outside of the yellow rectangle defined by  $a$  and  $b$ . We will make frequent use of this property of Manhattan metric.

In our discrete Manhattan space the pixel distance function  $mh$  is integer-valued, and the smallest steps we can take are of unit length in the directions N, W, S, E (North, West, South, East). Every pixel in our space has exactly four immediate neighbours, i.e. pixels reachable from it in one step. We will denote these neighbours of  $p$  by  $pix(p, N)$ ,  $pix(p, W)$ ,  $pix(p, S)$ ,  $pix(p, E)$ .

With these constructs in place, we can define an edge of an arbitrary but finite set  $A$  of Boolean-valued pixels in our discrete Manhattan space  $\mathbf{S}$ . A pixel  $a \in A$  belongs to an edge of  $A$  if at least one of its immediate neighbours does not belong to  $A$ , or, symbolically:

$$(5.5) \quad \text{edge}(A) = \{ a \in A \mid \text{val}(a) \text{ and } (\text{not}(\text{val}(pix(a, N)) \text{ and } \text{val}(pix(a, W)) \text{ and } \text{val}(pix(a, S)) \text{ and } \text{val}(pix(a, E)))) \}$$

The computation identifying  $\text{edge}(A)$  is not without cost, of course. This cost can be minimized with careful selection of data structures used to represent  $A$ . Our consideration of this issue will later follow.

We will find it useful to be able to inscribe arbitrary patterns within smallest possible rectangles. A rectangle in our Manhattan space can be defined by specifying its two diagonally opposite corners  $c_1$  and  $c_2$ .

Consider an arbitrary pattern  $A \subset \mathbf{S}$ . The smallest rectangle that may contain  $A$  is a set in  $\mathbf{S}$  defined as:

$$(5.6) \quad \text{rect}(A) = \text{rect}(\{c_1, c_2\}), \text{ where the pixels } c_1 \text{ and } c_2 \text{ have the following coordinates:}$$

$$c_1 = \langle \min(a.x \mid a \in A), \min(a.y \mid a \in A) \rangle$$

$$c_2 = \langle \max(a.x \mid a \in A), \max(a.y \mid a \in A) \rangle$$

Armed with these tools we strive to further reduce the computational effort needed to compute distance between arbitrary sets in discrete Manhattan space. Recall that on the basis of Theorem 2, having removed common pixels, we can limit our consideration to mutually exclusive sets in that space.

**Theorem 3:** For any two mutually exclusive patterns  $A, B \subset \mathbf{S}$  we have

$$\Delta(A, B) = \Delta(A, \text{edge}(B)).$$

**Proof:** Consider two points  $a \in A$  and  $b \in B$  maximizing the value of expression (3.2). That is,  $mh(a, b) = \Delta(A, B)$ . Could the point  $b$  lie in the interior of  $B$ ? That is, could  $b \in B \setminus \text{edge}(B)$ ? We will show that it is not possible.

Suppose that it were so that  $b \in B \setminus \text{edge}(B)$ . Then all four immediate neighbours of  $b$  would belong to  $B$ . One of those neighbours would be closer to  $a$  than  $b$  is. In that situation the two

points  $a$  and  $b$  could not maximize the value of the expression (3.2), contradicting the first statement of our proof.

From that it follows that  $b$  cannot be the interior point of  $B$ , therefore  $b \in \text{edge}(B)$ . But in this is the case  $\Delta(A, B) = \Delta(A, \text{edge}(B))$  holds. ■

This is a very useful finding, because it allows us to remove even more pixels from the interior of mutually exclusive patterns when computing  $MH(A, B)$  as in (5.2). In short, for mutually exclusive patterns  $A$  and  $B$  we can compute

$$(5.7) \quad MH(A, B) = \max \{ \Delta(A, \text{edge}(B)), \Delta(B, \text{edge}(A)) \}$$

instead of the formula (5.2), thus further reducing computing cost and time.

Encouraged by this success, we may attempt to compute

$$(5.8) \quad MH(A, B) = \max \{ \Delta(\text{edge}(A), \text{edge}(B)), \Delta(\text{edge}(B), \text{edge}(A)) \}$$

Will this always work? Unfortunately not. Consider, for example, a convex pattern  $A$ , entirely surrounded by pattern  $B$ . You will find that the pseudo-distance  $\Delta(A, B)$  will stretch from the interior of  $A$  to the edge of  $B$ . The following, however, will work:

**Theorem 4:** Given two non-empty patterns  $A$  and  $B$  such that  $\text{rect}(A) \cap \text{rect}(B) = \emptyset$ , the following holds:

$$MH(A, B) = \max \{ \Delta(\text{edge}(A), \text{edge}(B)), \Delta(\text{edge}(B), \text{edge}(A)) \}$$

**Proof:** Observe that the condition  $\text{rect}(A) \cap \text{rect}(B) = \emptyset$  implies mutual exclusivity of sets  $A$  and  $B$  but is stronger than that, i.e. this condition does not automatically hold for any pair of mutually exclusive sets  $A$  and  $B$ . Consider for example a rectangle in which two diagonally opposite corners constitute set  $A$ , and the remaining corners constitute set  $B$ . Both sets are mutually exclusive, but their rectangles are not.

Consider therefore two points  $a \in A$  and  $b \in B$  maximizing the value of expression (3.2). That is,  $mh(a, b) = \Delta(A, B)$ . We already know that  $b \in \text{edge}(B)$ . But now we ask: Could the point  $a$  lie in the interior of  $A$ ? That is, could  $a \in A \setminus \text{edge}(A)$ ? We will disprove this by contradiction, as in Theorem 3.

Suppose that  $a \in A \setminus \text{edge}(A)$ . It is possible then to execute one step of unit length from  $a$  in any direction and still remain within  $A$ . Given that  $a \notin \text{rect}(B)$  and  $B \subset \text{rect}(B)$ , at least one of these directions will lead away from  $\text{rect}(B)$ , thus leading away from  $B$ . Having executed one step from  $a$  away from  $B$  we find ourselves further from  $B$  than  $a$  was. Therefore our two points  $a \in A$  and  $b \in B$  could not maximize the value of expression (3.2). That is, for these points  $mh(a, b) \neq \Delta(A, B)$ , which contradicts our tentative assumption.

From that we conclude that  $a$  must belong to the edge of  $A$ , i.e.:  $a \in \text{edge}(A)$ . ■

Observe that the above theorem can be easily generalized: instead of covering patterns A and B with their rectangles, we may wish to cover them with arbitrary convex sets. The nature of the proof will remain the same. We use here the term “convex” in the Manhattan sense.

The above theorems constitute powerful tools, permitting us to ignore many pixels of A and B, when computing their similarity, thus vastly shortening the computational cost and time.

## 6. Coding notation and fundamental pseudocodes

Until now we were successfully reducing the computational complexity of the problem at hand by removing as many pixels as possible from patterns under consideration. Now we focus on reducing computational time, by exploit massive parallelism inherent in our formulae. The hardware and software of today is not entirely suitable for our purpose, so we resort to the following pseudocode:

### 6.1. Pseudocode notation

We use standard notation, enhanced only in order to express parallelism and mutual exclusion to protect write access to shared variables. The notation

```
parbegin
  statement1;
  statement2;
  statement3;
parend;
statement4;
```

means that statements 1, 2 and 3 are to be executed in parallel, and the execution of statement 4 may begin only after all the three preceding statements in the **parbegin** block have terminated. Each of these three statements may be simple or a composite (i.e. a block). A composite statement is made of several statements (optionally preceded by declarations) and enclosed within **begin** and **end**.

For example, let us write the pseudocode of the parallel function computing Euclidean distance between points in a 3D space. In this space each point  $P = \langle x, y, z \rangle$  is an ordered triple, where  $x, y, z$  are the coordinates of  $P$ . In the pseudocode the notation  $P.x$  means the  $x$ -coordinate of point  $P$ .

```
function d1(P1, P2 : Point) return real
is
  disx2, disy2, disz2 : real;
begin
  parbegin
    disx2 := square(P1.x - P2.x);
    disy2 := square(P1.y - P2.y);
    disz2 := square(P1.z - P2.z);
  parend;
  return sqrt(disx2 + disy2 + disz2);
end d1;
```

Several statements executing in parallel may attempt to modify some shared variables. To ensure integrity of such data we will enforce mutual exclusion of write access by using semaphores. A

semaphore can store only non-negative values and can be initialized to any such value. Only two operations on a semaphore are possible:

```
signal(s); -- increments the value of semaphore s by 1
wait(s);   -- decrements the value of semaphore s by 1 as soon as
           -- it is possible to do so without yielding s negative
```

More info on handling concurrency issues, such as access to shared variables, use of mutual exclusion and semaphores can be found in [8], [9].

Using a mutual exclusion semaphore we may provide conceptual pseudocode for the metric (2.7), now, as an example, between points in a 3D space:

```
function d3(P1, P2 : Point) return real
is
  distance : real := 0;
  mutex : semaphore := 1; -- semaphore initialized to 1
begin
  parbegin
    begin disx : real := abs(P1.x - P2.x);
      wait(mutex);
      if distance < disx then distance := disx; end if;
      signal(mutex);
    end;
    begin disy : real := abs(P1.y - P2.y);
      wait(mutex);
      if distance < disy then distance := disy; end if;
      signal(mutex);
    end;
    begin disz : real := abs(P1.z - P2.z);
      wait(mutex);
      if distance < disz then distance := disz; end if;
      signal(mutex);
    end;
  parend;
  return distance;
end d3;
```

In this solution some sections of the code execute in parallel, but updating of the shared variable `distance` is protected by mutual exclusion. The three updates can be performed in arbitrary order. The scope of variables `disx`, `disy`, `disz` and of the semaphore `mutex` is the block within which they are declared, and its inner blocks.

In our pseudocodes we will also use the `parfor` statement. The notation

```
parfor i in 1..10 do
  worker(i);
parend;
statement;
```

represents simultaneous launching of ten instances of the procedure `worker`, each with its own argument. It resembles the familiar `for` loop, but here the “iterations” are executed in parallel. As before, the execution of `statement` at end of this pseudocode may begin only after all ten worker processes in the `parfor` block have terminated.

Finally, the familiar **return** statement is used to immediately terminate the execution of the procedure or function containing it, while returning the execution control to the calling environment. Additionally it terminates all parallel execution processes that were created as a result of calling the procedure or function. If used within a function, it also returns its computed value. As an example of use see pseudocode (9.8).

Observe that our pseudocodes are perfectly suitable for execution on a single-core CPU. The **parfor** block reduces then to the usual **for** loop, while the compound statements within a **parbegin** block can be executed in any order.

## 7. Pseudocodes for computing distances between sets

For reasons to follow, we will use integer-based metrics. The examples below reflect this already.

The digital implementation of the set  $A$  being the argument of the function  $\delta(x, A)$  defined in (3.1) is a finite set of  $n$  points

$$(7.1) \quad A = \{ a_1, a_2, \dots, a_n \}$$

Using an arbitrary function  $d$  for computing distances between points, we can write the following pseudocode for computing  $\delta(x, A) = \inf \{ d(x, a) \mid a \in A \}$  defined in (3.1):

```
(7.2) function  $\delta(x : \text{Point}; A : \text{Set})$  return integer
is
    distance : integer :=  $\infty$ ; -- initialized to infinity
    mutex    : semaphore := 1;
begin
    parfor  $i$  in 1 .. card(A) do -- card(A) is the number of elements in A
        dis : integer :=  $d(x, a(i))$ ;
        wait(mutex);
        if  $dis < distance$  then  $distance := dis$ ; end if;
        signal(mutex);
    parend;
    return distance;
end  $\delta$ ;
```

We are now ready to define the pseudocode for  $\Delta(A, B) = \sup \{ \delta(a, B) \mid a \in A \}$  given in (3.2) as a pseudo-distance between two sets  $A = \{ a_1, a_2, \dots, a_n \}$  and  $B = \{ b_1, b_2, \dots, b_m \}$ :

```
(7.3) function  $\Delta(A, B : \text{Set})$  return integer
is
    distance : integer := 0;
    mutex    : semaphore := 1;
begin
    parfor  $i$  in 1 .. card(A) do -- card(A) is the number of elements in A
        dis : integer :=  $\delta(a(i), B)$ ;
        wait(mutex);
        if  $dis > distance$  then  $distance := dis$ ; end if;
        signal(mutex);
    parend;
end  $\Delta$ ;
```

```

    parent;
    return distance;
end  $\Delta$ ;

```

Finally the distance  $D(A, B)$  between two sets defined in (3.3) can be implemented simply as:

```

(7.4) function D(A, B : Set) return integer
is
    disAB, disBA : integer;
begin
    parbegin
        disAB :=  $\Delta(A, B)$ ;
        disBA :=  $\Delta(B, A)$ ;
    parent;
    return max(disAB, disBA);
end D;

```

Following this logic, an arbitrary function  $d$  used to measure distance between points of a metric space has the properties (2.1), (2.2), (2.3) and (2.4) and induces another function  $D$  that can be used to measure distances between non-empty subsets of that space. That new function  $D$  also meets the requirements (2.1), (2.2), (2.3) and (2.4).

Furthermore, if the function  $d$  can be implemented on a digital computer, then the function  $D$  can also be implemented, as per pseudocodes (7.2), (7.3) and (7.4).

In our further analysis we therefore will only need to show how to implement the function  $d$  on a digital computer. That function induces the existence of the function  $D$ .

## 8. On comparison of patterns

Pattern recognition and classification relies on routine use of pattern comparison algorithms. Prior to any comparison, two patterns must be optimally aligned; any residual distance between two patterns is then a measure of their dissimilarity. Two patterns are deemed similar enough if their dissimilarity falls within tolerances of an application.

The optimal alignment of patterns involves a linear transformation of one pattern with regard to another minimizing the distance between two patterns. This linear transformation may involve translation, rotation and scaling. The nature of the application dictates which of these three operations is admissible.

For example: When comparing pair-wise the signatures in Fig.1 we might want to slide one signature onto another, in order to focus our attention on dissimilarities. Translation is therefore allowed here. Is scaling allowed? Probably not. After all, we (humans) tend to sign documents in a similar way, regardless how much space for the signature is left on a document. Certainly rotation is not allowed: a signature oriented upside-down or at  $90^0$  angle would raise serious doubts about its validity.

Similarly, humans are experts at face recognition, but not when faces are presented to us upside-down. In such situation we are unable to recognize even members of our immediate families.

Consequently, prior to comparing any two patterns A and B we will assume that they are optimally aligned via pre-processing suitable to the application at hand. In any case, this pre-processing will include removal of common pixels (thus making A and B mutually exclusive) and, whenever allowed, the interior points of A and B.

### 8.1. Search strategies: Basic techniques for low-resolution patterns

Analytically speaking the problem of comparing two optimally aligned patterns has already been solved: given two patterns A, B one needs to compute the distance  $D(A, B)$  between them as per pseudocodes (7.2), (7.3) and (7.4). If both sets A and B are relatively small (say, originally consist of a small number of N elements, further reduced by pre-processing), we could implement A, B as lists. The performance of the algorithm implementing the distance  $D(A, B)$  would be  $O(N^2)$ . Such delay is acceptable for small N, i.e. for low-resolution patterns.

Indeed, things could be accelerated further by refining the implementation of the function  $\delta(x, A) = \inf \{ d(x, a) \mid a \in A \}$  defined in (3.1). A better implementation of (6.2) would be:

```
(8.1) function  $\delta(x : \text{Point}; A : \text{Set})$  return integer
is
    distance : integer :=  $\infty$ ; -- initialized to infinity
    mutex    : semaphore := 1;
begin
    parfor i in 1 .. card(A) do -- card(A) is the number of elements in A
        dis : integer := d(x, a(i));
        if dis = 1 then return 1; end if;
        if dis < distance then
            wait(mutex);
            if dis < distance then distance := dis; end if;
            signal(mutex);
        end if;
    parend;
    return distance;
end  $\delta$ ;
```

Observe that the function  $\delta(x, A)$  scans all points in A looking for the one closest to x. Should it find the point in A distant by 1 from x, it immediately returns the value 1, which is the minimum value possible for mutually exclusive sets A and B. There is no need to peruse the remainder of points in A, which instantly become “uninteresting”. There is no gain in speed, if we can execute the **parfor** loop in perfect parallelism. Should we have fewer than  $\text{card}(A)$  processors, the speed gain can be noticeable. Indeed, the smaller the number of processors, the more noticeable it becomes.

Optimal alignment of patterns A and B increases the probability of some points being immediate neighbours. The more of them, the faster our algorithm becomes. In principle, however, there is no guarantee that any points would be immediate neighbours: in that case the computation of the distance  $D(A, B)$  using (7.2) instead of (8.1) would be faster. Indeed, when pressed for time, we should evaluate this distance twice and in parallel, using implementations (7.2) and (8.1). The computation of distance  $D(A, B)$  terminates whenever its result is returned first by one of the two parallel branches.

Observe further that there is no need for every parallel instance of the `parfor` block in the pseudocode (8.1) to update the shared variable `distance`, which holds the best distance value from  $x$  to  $A$  seen so far. The shared variable is protected by the semaphore `mutex`, which enforces mutual exclusion of access to update that variable. This mutual exclusion, when applied to all instances of the `parfor` block would amount to code serialization, resulting in a serious execution slow-down. Peruse the highlighted section of the pseudocode (8.1). The mutually exclusive code is put within an `if` statement, which inspects the value of the `distance` variable in the insecure way. Only when the chances of meaningful update are reasonably high, the critical section of the code is executed.

The same execution acceleration technique can be used when implementing the pseudocode for  $\Delta(A, B) = \sup \{ \delta(a, B) \mid a \in A \}$  given in (3.2) as a pseudo-distance between two sets  $A = \{ a_1, a_2, \dots, a_n \}$  and  $B = \{ b_1, b_2, \dots, b_m \}$ :

```
(8.2) function  $\Delta(A, B : \text{Set})$  return integer
is
    distance : integer := 0;
    mutex    : semaphore := 1;
begin
    parfor i in 1 .. card(A) do -- card(A) is the number of elements in A
        dis : integer :=  $\delta(a(i), B)$ ;
        if dis > distance then
            wait(mutex);
            if dis > distance then distance := dis; end if;
            signal(mutex);
        end if;
    parend;
    return distance;
end  $\Delta$ ;
```

Finally the distance  $D(A, B)$  between two sets defined in (3.3) can be implemented simply as per pseudocode (7.4).

## 8.2. Search strategies: Refined techniques for higher-resolution patterns

We concern ourselves here with computing efficiently the value  $\delta(x, A) = \inf \{ d(x, a) \mid a \in A \}$  (recall (3.1)). We have already made some steps in this direction: the eighth line in the pseudocode (8.1) reads:

```
if dis = 1 then return 1; end if;
```

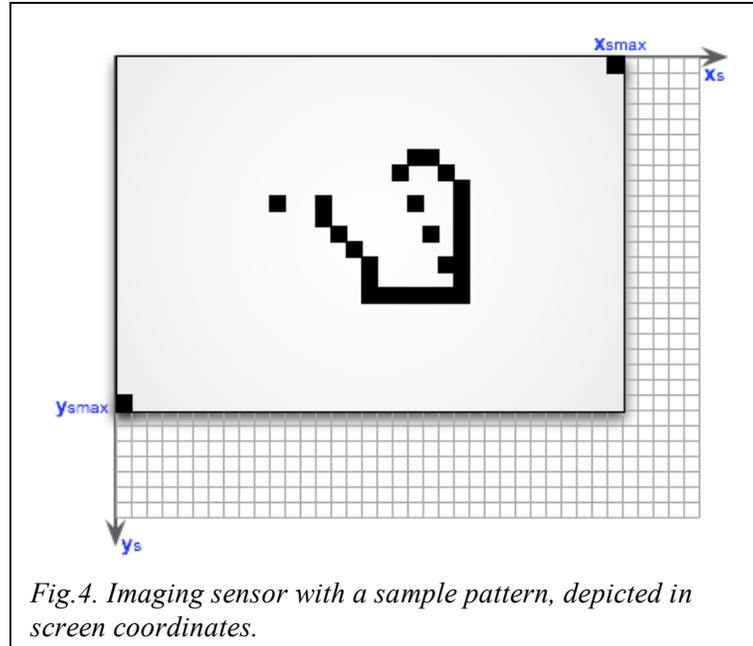
The value 1 is the lowest value the variable `dis` may assume, given that common points were removed from patterns  $A$  and  $B$ . Should this value be encountered, there is no point in searching for a better value. The sooner we can terminate the search, the better. Indeed, we already have done some improvements to this search method: Prior to removal of common points from patterns  $A$  and  $B$  the lowest possible search value was 0. Now, with the new value of 1 we have roughly quadrupled the probability of terminating the search early.

We would like to be able to interrogate our pattern data for such critical values first, instead of waiting for our luck to encounter these critical values. To that end we will retain the structure of a

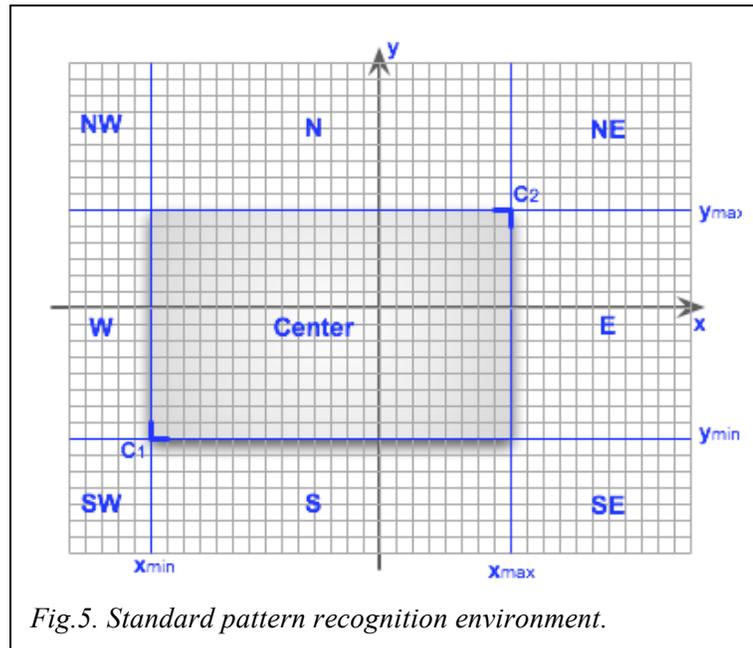
pixel list, but will place the elements of the lists, being the info regarding pixels of patterns under consideration, within the discrete Manhattan pixel space  $\mathbf{S}$ .

The very structure of current imaging sensors facilitates this task: Fig.4 illustrates the organization of a typical imaging sensor (usually a CCD or CMOS device), capturing a sample pattern, shown here in the “screen coordinates”, traditionally used in computer graphics.

Such sensors are actually  $x_{smax}$  by  $y_{smax}$  rectangular arrays of pixel sensors. In our example the pattern captured contains two pixels defining the sensor size. Normally, however, the captured pattern is contained inside the sensor area. Should our pattern of interest fail to fit the sensor region of the space  $\mathbf{S}$ , we can take a finite number of partially overlapping images of it, and digitally stitch them together to form one composite image. Such digital stitching of a finite number of photo shots is done frequently today. Without any loss of computability we can therefore assume that any pattern of our interests is finite and fits some imaging sensor.



While inspecting pre-processed patterns for the purposes of pattern recognition we prefer to operate within the traditional systems of coordinates. Fig. 5 depicts our environment for pattern recognition. The axes  $Ox$  and  $Oy$  are oriented in a standard way. The central rectangle contains the smallest region of the imager containing a pattern in question. This central rectangle is surrounded by eight zones labeled N, NE, E, SE, S, SW, W and NW, which together tile the pixel space  $\mathbf{S}$ .



Our pattern (for simplicity not drawn in Fig.5) is positioned so that its pre-processed reference point is at the origin of the system of coordinates. As indicated before, the kind of pre-processing needed depends on the nature of the application.

For example, in the signature authentication illustration provided later in the paper, we align signature patterns so that their centers of mass coincide at origin. In signature pre-processing only linear shifts were allowed, but no scaling nor rotation. The scanned signatures were flipped around  $Ox$  axis to reflect the standard orientation of the  $Oy$  axis.

Note that this data structure allows us to pre-process the pattern data with greater efficiency. Removal of common pixels from two patterns can be now done in  $O(N)$  steps, where  $N$  is the cardinality of the smaller of the two patterns: for each pixel in the first pattern we take its  $\langle x, y \rangle$  coordinates and check whether the second pattern has a corresponding pixel. If so, then we remove the shared pixel from both patterns.

Similarly, removal of the interior pixels from a given pattern is fast and easy: knowing the  $\langle x, y \rangle$  coordinates of each pixel in a pattern we can immediately check whether all its immediate neighbours belong to that pattern: if so, that pixel can be removed.

With complex and similar patterns the probability of point proximity increases. We will take advantage of this. Having removed common pixels, we can now check for points in a given pattern distant by 1 from  $x$ . Should we find one, then the function  $\delta(x, A)$  would immediately return the value 1, without wasting time to inspect the remainder of the pattern  $A$ . Failing that, we would check for points distant by 2, 3, ... etc. from  $x$ . This can be done by inspecting concentric circles of radii 1, 2, 3, etc. centered at point of interest  $x$  (peruse Fig.6). Observe that this algorithm always stops because our pattern being inspected is finite but non-empty and fits within its own rectangle (see Fig.5), defined by the points  $C_1 = (x_{\min}, y_{\min})$  and  $C_2 = (x_{\max}, y_{\max})$ .

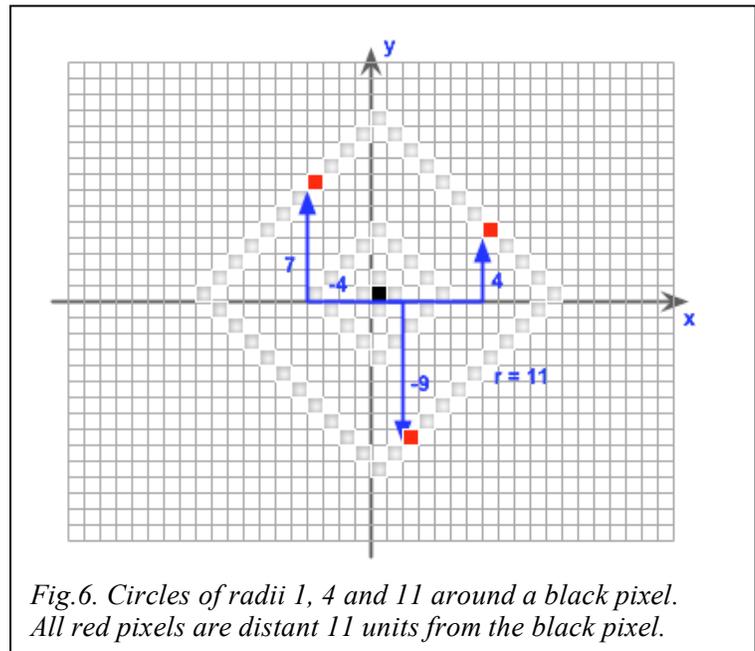


Fig.6. Circles of radii 1, 4 and 11 around a black pixel. All red pixels are distant 11 units from the black pixel.

Note further that circles in Manhattan space (again see Fig.6) have “sides” (recall the ball  $d_2$  in Fig.2), which we will label NE, NW, SE, SW (North East, North West, etc.) The circumference of such a circle is  $8 \times \text{radius}$ . Each side, of length  $2 \times r$ , is made of  $r$  pixels only. Traversal of each side is easy: we merely need to keep modifying one coordinate (increasing or decreasing it by 1) while respectively modifying another coordinate, repeating that modification (preferably in parallel)  $r$  times. Even on a parallel pattern-matching machine with unlimited number of processors it does not make much sense to scan a circle of radius  $r+1$  until a circle of radius  $r$  is inspected. However, all pixels belonging to a “side” of a circle can be inspected in parallel. Similarly, the four sides of a given circle can be inspected in parallel as well.

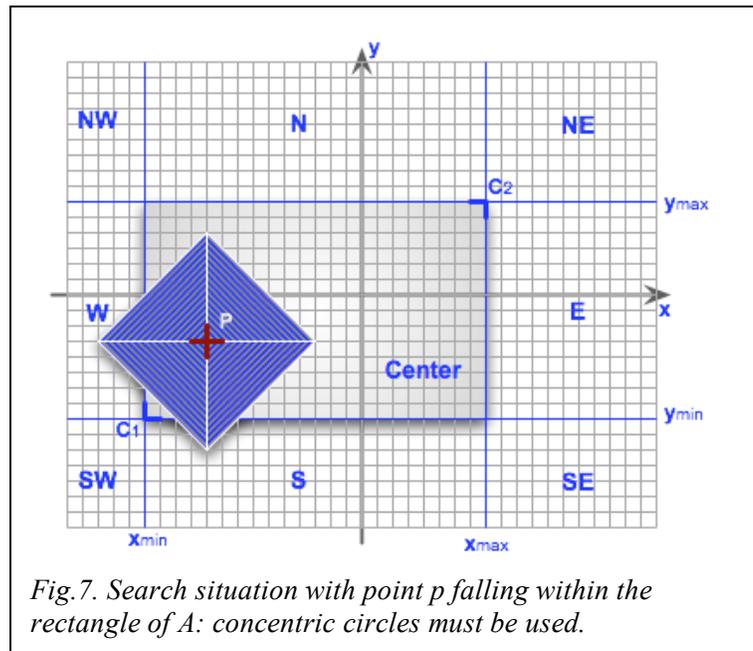
As an interesting aside, please note that in our discrete Manhattan pixel space  $\mathbf{S}$  the areas of circles can, strangely enough, be measured in pixels too. We leave to the reader to find the formula to find the area of such a circle, as an exercise.

To emphasize that we are dealing with pixel data, we will be writing  $\delta(p, A)$  the, where  $p$  is a pixel, and  $A$  is a set of pixels. The specification of the function  $\delta(p, A)$  becomes:

(8.2) `function`  $\delta(p : \text{Pixel}; A : \text{Set})$  `return` integer;

As mentioned before, the point  $p$  may lay in the central rectangle of pattern  $A$ , or in any other zone adjacent to that rectangle (as per Fig.5). This is so, because  $p$  belongs to another pattern we are comparing to  $A$ . These patterns not being necessarily identical may lie within two rectangles that need not be congruent.

Most frequently the point  $p$  will fall within  $\text{rect}(A)$ . In search for a point  $a \in A$  closest to  $p$  we have to draw concentric circles around  $p$  until one of them includes some  $a \in A$ . That search situation is depicted in Fig.7. The search terminates as soon as the first such point  $a \in A$  is found. Note that as the radius of our search grows, some “sides” of our circles may lay (partially or entirely) outside of  $\text{rect}(A)$ . We need to inspect only the parts of the sides lying within  $\text{rect}(A)$ , knowing that there are no pixels of  $A$  outside  $\text{rect}(A)$ . Given that  $\text{rect}(A)$  is convex (in the Manhattan sense), there can be at most one segment of each side within  $\text{rect}(A)$ . Knowledge of these facts permits us to write very efficient search code.



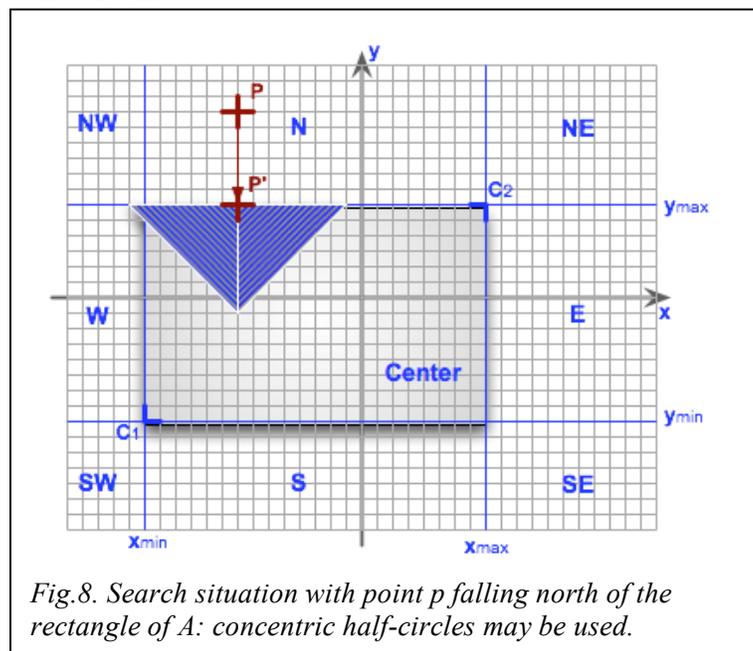
Sometimes the point  $p$  may fall into one of the zones N, S, E, W adjacent to  $\text{rect}(A)$ . In this situation we project  $p$  onto the rectangle yielding  $p'$  and we are inspecting half-circles until we find  $a \in A$  closest to  $p'$ . (See Fig.8). Then we compute the distance:

$$(8.3) \quad d(p, a) = d(p, p') + d(p', a)$$

because the points  $p$ ,  $p'$  and  $a$  are co-linear (in the Manhattan sense), i.e. the point  $p'$  belongs to the rectangle defined by corners  $p$  and  $a$  (recall reasoning depicted in Fig.3).

Whenever point  $p$  falls into one of the corner zones NE, NW, SE, SW (see Fig.9) we will denote as  $p'$  the relevant corner of the rectangle and will limit ourselves to drawing only quarter-circles until we find  $a \in A$  closest to  $p'$ . Then again we will compute the distance  $d(p, a)$  using formula (8.3).

### 8.2.1. We can compute it even faster!

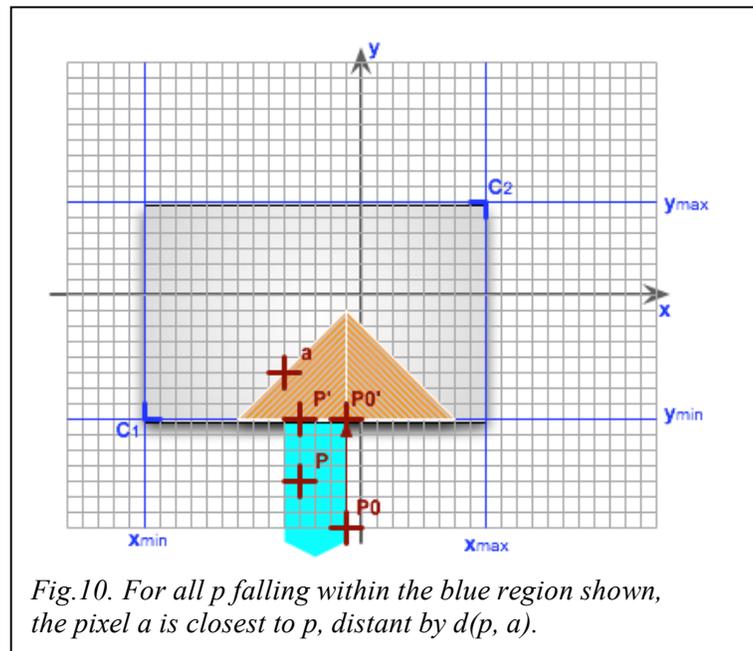
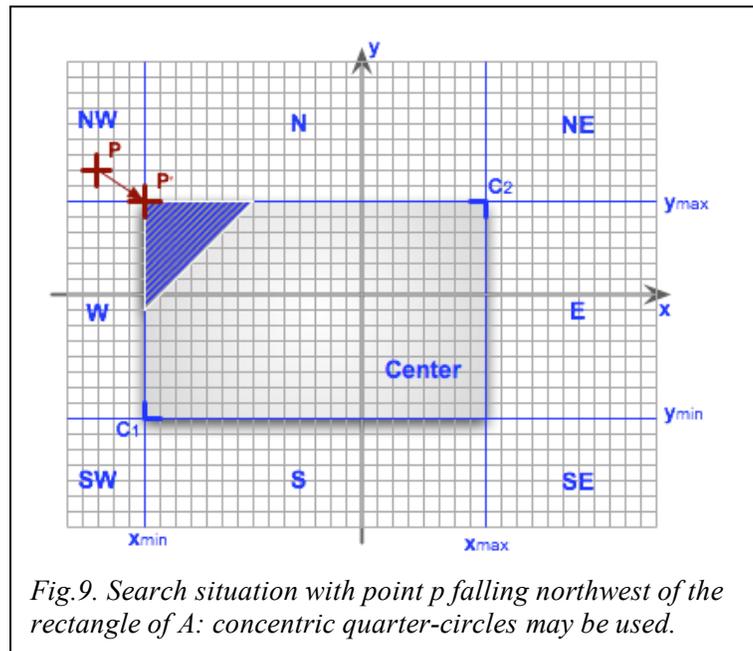


Being able to search by inspecting half- or quarter-circles only in relevant situations can accelerate computations considerably. We can do even better, though. Observe that **for all pixels  $p$  in the corner zones NE, NW, SE, SW** (see again Fig.9) the term  $d(p', a)$  in the formula (8.3) will yield the same result for all  $p$  in each corner zone. Consequently, for each corner zone such search should be conducted only once and its outcome remembered.

**For the remaining N, S, E, W zones** analogous (although a bit more complex) short-cuts are possible. As an illustration, consider the situation concerning zone S depicted in Fig.10. Suppose that previous search succeeded in finding  $a \in A$  closest to  $p_0$ . Suppose further that now we seek  $a \in A$  closest to  $p$ . Observe that  $p'$  belongs to the rectangle defined by points  $a$  and  $p_0'$ , therefore the distance between  $p$  and  $a$  can be computed simply as (8.3). Why? Given that  $a$  is closest to  $p_0'$  and  $p'$  is closer to  $a$  than  $p_0'$ , then  $a$  must be closest to  $p'$ . In fact, for all  $p$  falling onto an infinitely tall blue rectangle shown on Fig.10 the formula (8.3) applies; no search is needed.

Similarly, for all  $p$  falling within the rectangle defined by  $a$  and  $p_0$  (not shown on Fig.10) we can compute the distance  $d(p, a)$  directly, without any search.

This finding gives us a useful tip: when inspecting sides of the circles our algorithms should report the positions of all pixels found, rather than merely reporting that the search was successful. These positions might be found useful later, obviating the needs for some subsequent searches. For all other  $p$  in zone S, such that  $p.x < a.x$  or  $p.x > p_0.x$  there is no guarantee that  $a$  is closest to  $p$ , but the prior knowledge of existence of  $a$  allows us to estimate the upper limit of the computational effort needed to find a better  $a$  (if any). This is useful in processor scheduling. Of course that search should avoid inspecting regions already found to be empty (shown in dashed orange).



The logic concerning zone S, presented as an illustration here, applies to zones N, E, W, with obvious geometrical modifications.

**For  $p$  falling in the center zone** we can find similar computational shortcuts. For example: for all  $p$  falling onto the blue rectangle defined by  $a$  and  $p_0'$  (see Fig.11), the pixel  $a$  is closest to  $p$ , the distance being  $d(p, a)$ . No search is needed.

Similar situation arises when  $p$  falls close to a corner of the center zone. Consider the previous search situation depicted in Fig. 9, which resulted in finding some point  $a \in A$  (not shown). We can envisage a rectangle defined by that  $a$  and  $p'$ . Should any new  $p$  fall onto that rectangle, then  $a$  is closest to that  $p$ . No need to search.

Finally, consider the situation in which a previous search called for using full circles (Fig.12). The previous logic still applies.

Specifically: Suppose the former search for  $a \in A$  closest to  $p_0$  resulted in finding  $a$  as shown. Then, for any new  $p$  falling onto the blue rectangle defined by  $p_0$  and  $a$  there is no need to perform any search: we already know that the same  $a$  is still relevant.

Remember further that any search, like the one shown in Fig.12, may result in several  $a$  found, all equidistant from its origin. In that situation several blue rectangles can be drawn. Should a new search be required then the first thing to do is to check whether the new origin falls onto any one of the blue rectangles.

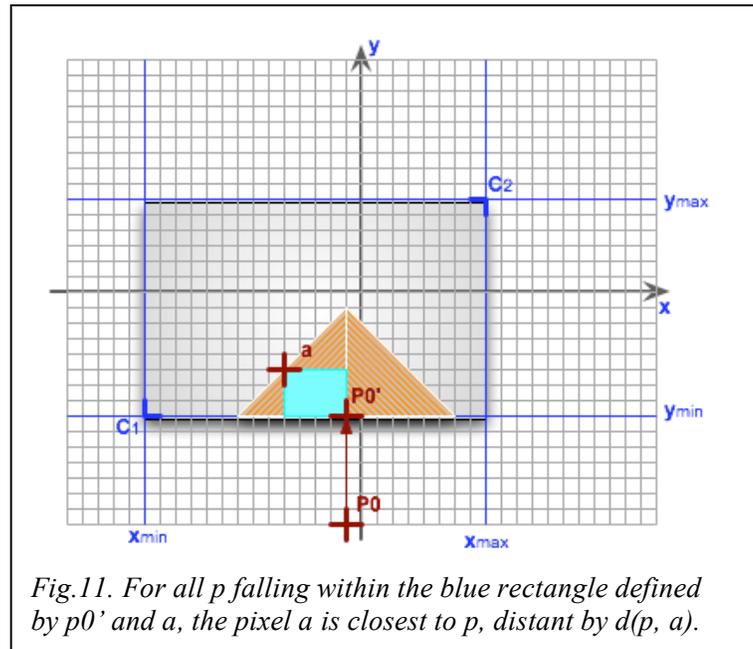


Fig.11. For all  $p$  falling within the blue rectangle defined by  $p_0'$  and  $a$ , the pixel  $a$  is closest to  $p$ , distant by  $d(p, a)$ .

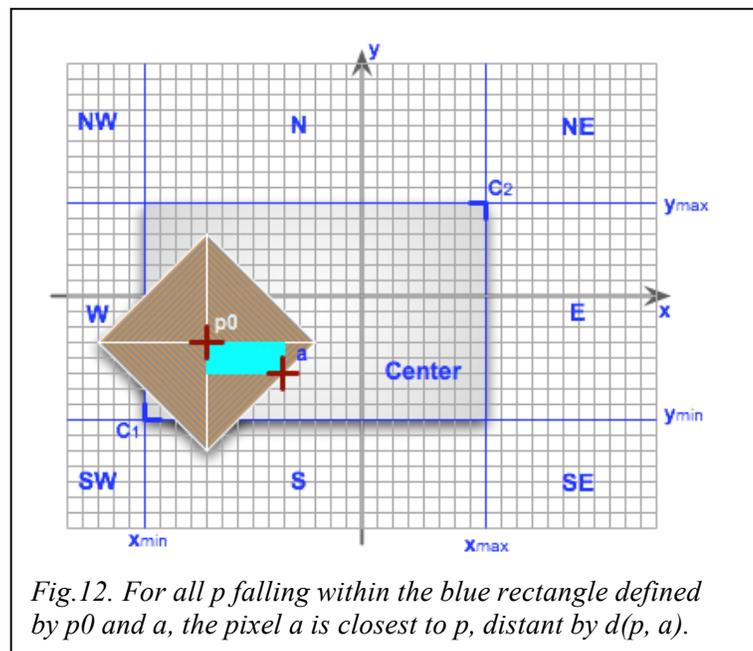


Fig.12. For all  $p$  falling within the blue rectangle defined by  $p_0$  and  $a$ , the pixel  $a$  is closest to  $p$ , distant by  $d(p, a)$ .

Clearly, the information about “voids” within a scanned pattern is valuable, as it can be used to avoid performing unnecessary, repetitive searches. An open question arises, calling for future investigation: Under what conditions an additional pre-processing of patterns aimed at discovery of these voids would be beneficial? By the nature of things, patterns do not fill their rectangles completely – this means that some voids are initially there. Such voids are further expanded (and new voids created) by removing shared pixels from patterns, and by removing (whenever possible)

the interior pixels from patterns. When would a pre-processing exercise of void identification by inscribing a number of largest possible empty circles within a pattern be beneficial? How many of such empty circles should we attempt to inscribe? What is the minimum radius of a circle worth inscribing? The answers to these questions depend on the nature of patterns (i.e. the environment the system inhabits), the resolution of their imaging sensors and on the computing performance characteristics of the system (i.e. the degree of adaptation of the system to its environment).

This concludes our outline of strategies applicable to accelerating search computations when using Manhattan metric. Observe that ours is a gross simplification of the biological vision: animal retinae contain sensors arranged in a hexagonal grid. In that situation the pixel space can be subdivided into 19 zones!

## **9. Application example: Signature authentication**

To illustrate the power of our basic strategies we looked for a practical problem of pattern recognition, clustering and classification that would be of universal interest. We chose an everyday problem of signature authentication. Signatures are relatively simple patterns, and in their study we can skirt a number of pre-processing issues of computer vision: extraction of patterns from their backgrounds, pattern translation, rotation and scaling, handling of image noise, of colour, of contrast and exposure, etc. Our approach permits us to handle these important issues, but this is a topic for another paper.

Despite of its simplicity, the problem of signature authentication is of immediate practical applicability and interest. Or perhaps its simplicity is deceptive?

Signatures are routinely used in banking, for example. Banking is a discipline older than computer science, so surely bankers must have developed over time some standard methods for signature authentication... We wanted then to compare the reliability of our methodology to a reliability of a trained expert banker (a human), treated as a control. Unfortunately, we could not find any study evaluating the training effectiveness of bank clerks in signature authentication, or identification of so skilled clerks. In fact, oddly it seems that the bankers do not attempt to verify signatures on all documents passing through their hands, but merely look at signatures only on documents that are found problematic otherwise.

If in doubt, consider a clerk in a bank where you have an account. That clerk is responsible of serving several hundred (perhaps more?) customers like you. Each of these people cannot sign twice exactly in the same way while the forgeries of their signatures are likely to resemble genuine signatures. What is the chance that your clerk intricately knows the way you pen your signature, as well as the penmanship of all his other customers, to reliably judge each signature presented to him as genuine or a fake?

Facing this reality we cannot offer a traditional, comparative study of our signature authentication methodology vs. that of a trained human, treated as a control. We can, however, present the reliability of our system in absolute terms.

To that end we gathered the signatures of 21 persons, whom we knew well. This is too small a sample to offer statistically firm findings but only 21 persons well known to us agreed that we use their signatures in our study. We then decided to focus on the trends only.

Each participant in our study deposited a signature eight times (four times when “opening an account” with us, and four more times simulating subsequent “transactions”). We (the authors) also forged some of their signatures, and also asked some randomly chosen participants to forge some signatures of other participants. In this way for each participant we had 12 signatures: 8 genuine and 4 forgeries. We could tell for sure which is which, because we witnessed their creation. Thus, to test our system we had a pool of 252 signatures in total.

*A priori*, our system can recognize a valid signature as such, or a forgery as a fake. In both cases it is deemed successful. It can also commit type I error (judging a valid signature as fake) or type II error (diagnosing a forged signature as genuine). In the latter two cases the system would be unsuccessful. The overall reliability of the system is the probability of it being successful when presented a sequence of signatures. This allows us to measure the reliability in absolute terms; moreover, it allows conducting simulation studies of account handling procedures that would enhance banking security. Note that the bankers are acutely concerned with type II errors only: these errors make them lose money. The errors of type I merely annoy the customer, who on second thought may actually congratulate a banker on his vigilance.

Using this theory we can offer the methodology for automated signature authentication, as our application example. In current banking procedures, both signature and customer name appear on a document, and the signature is merely used for a visual verification by a bank clerk that a given customer authored the document. That document may bear a correct mailing address of the sender and other information facilitating the signature validation task, which is simpler than the problem of customer identification given signature only (or a fingerprint, a photograph, etc.), but still error-prone.

Our system is more ambitious: it attempts to identify the customer on the basis of the signature alone. It compares signatures to reference signatures. Each signature  $S$  is scanned in black and white at 300 dpi resolution and is stored as a set  $S = \{p_1, p_2, \dots, p_n\}$  of black pixels  $p_i = \langle x_i, y_i \rangle$  where  $x_i, y_i$  are integer-valued pixel coordinates. Different signatures consist of varying number and positions of black pixels. We assume here that all signatures in the system are preprocessed so that their relative positioning information is deliberately lost: we made their centers of mass coincide. Neither signature rotation nor scaling was allowed.

We use the integer-valued Manhattan metric (2.6) to measure distance between pixels. This induces a particular metric  $D$  (as per 3.3) to be used for measuring distance between signatures. With the positioning information lost,  $D$  can be seen as a measure of dissimilarity between signatures. In this vein two signatures  $S_1$  and  $S_2$  are deemed to belong to the same person if they are sufficiently similar, i.e.  $D(S_1, S_2) \leq \epsilon$  where  $\epsilon$  is a properly chosen small value.

### 9.1. Opening an account in the traditional way

A new customer is requested to provide four reference signatures  $S_1, S_2, S_3, S_4$  that are certified on the spot by a witnessing clerk. We made the system open an account in three ways: using two, or three, or four signatures of a given customer. Using these balls the system create a new customer

signature profile C consisting of balls that partially overlap. The customer profile consisting of three balls follows:

$$(9.1) \quad C = \{ B_1(S_1, r_1), B_2(S_2, r_2), B_3(S_3, r_3) \} \text{ where}$$

$$r_1 = \min ( D(S_1, S_2), D(S_1, S_3) )$$

$$r_2 = \min ( D(S_2, S_1), D(S_2, S_3) )$$

$$r_3 = \min ( D(S_3, S_1), D(S_3, S_2) )$$

The pseudocode for creating this signature profile is as follows:

```
(9.2) procedure New_Profile(C : out SigProfile)
is
    B1, B2, B3 : SigBall;
begin
    C :=  $\emptyset$ ; -- makes C empty
    Read B1.S;
    Read B2.S;
    Read B3.S;
    parbegin
        B1.r := min(D(B1.S, B2.S), D(B1.S, B3.S));
        B2.r := min(D(B2.S, B1.S), D(B2.S, B3.S));
        B3.r := min(D(B3.S, B1.S), D(B3.S, B2.S));
    parend;
    Insert B1, B2, B3 into C;
end New_Profile;
```

Observe that one reference ball may contain another ball, making the smaller ball redundant. To remove redundant balls we may sort the balls in the signature profile  $c$  in descending order of the ball radius and check smaller balls if they are inside a larger ball. A ball  $B_j$  is inside  $B_i$  when

$$(9.4) \quad D(S_i, S_j) \leq r_i - r_j$$

Does it matter if some reference balls are removed? This may depend on business procedures. In any case, we may ask the customer to provide more signature samples if necessary. In our study we had to contend only with signatures we gathered *a priori*. Collection of extra signatures was not possible.

## 9.2. Opening an account in a novel way

As before, a new customer provides four reference signatures  $S_1, S_2, S_3, S_4$  that are certified by a witnessing clerk, who then immediately provides two forged signatures  $F_1, F_2$  instead of giving the criminals the opportunity to do so first. Again, we made the system open an account in three ways: using two, or three, or four signatures of a given customer, but always the two forgeries. As before, we made the system create a new customer signature profile C viz.:

$$(9.5) \quad C = \{ B_1(S_1, r_1), B_2(S_2, r_2), B_3(S_3, r_3) \} \quad \text{but now, given the forgeries we compute}$$

$$r_1 = \min ( D(S_1, F_1), D(S_1, F_2) ) - 1$$

$$r_2 = \min ( D(S_2, F_1), D(S_2, F_2) ) - 1$$

$$r_3 = \min ( D(S_3, F_1), D(S_3, F_2) ) - 1$$

In short, we make the balls as large as possible, while making sure that they do not touch (nor contain) the forged signatures. Recall that our function for measuring distance between pixels (2.6) was integer-valued, therefore the function  $D$  is integer-valued too. By subtracting 1 in the radii formulae above we make sure that the forgeries lay outside of the balls.

The pseudocode for creating a signature profile together with a forgery profile for a given customer is as follows:

```
(9.6) procedure New_Profile(C : out SigProfile; FSet : out ForgedSet)
is
    B1, B2, B3 : SigBall;
    F1, F2     : Signature;
begin
    C :=  $\emptyset$ ;      -- makes C empty
    FSet :=  $\emptyset$ ; -- makes FSet empty
    Read B1.S; Read B2.S; Read B3.S;
    Read F1;   Read F2;
    parbegin
        B1.r := min(D(B1.S, F1), D(B1.S, F2)) - 1;
        B2.r := min(D(B2.S, F1), D(B2.S, F2)) - 1;
        B3.r := min(D(B3.S, F1), D(B3.S, F2)) - 1;
    parend;
    parbegin
        Insert B1, B2, B3 into C;
        Insert F1, F2 into FSet;
    parend;
end New_Profile;
```

As before, the bankers might want to replace redundant balls in the signature profile with more relevant balls. In our study this was not possible, all the signature having been collected upfront. Additionally, should some radii of the reference balls be deemed too small, the banker may advise the customer that her/his signatures can easily be falsified.

### 9.3. Authenticating signatures

To be deemed valid each new signature of our customer must fall within at least one of the reference balls. A new signature  $X$  falls within a ball  $B_i(S_i, r_i)$  when

$$(9.7) \quad D(S_i, X) \leq r_i$$

therefore the authentication pseudocode is simply:

```
(9.8) function Valid(X : Signature; C : SigProfile) return Boolean
is
begin
    parfor i in 1 .. card(C) do
        if D(X, C.B(i).S)  $\leq$  C.B(i).r then return True; end if;
    parend;
    return False;
end Valid;
```

## 9.4. Learning from errors

No intelligent system, biological or otherwise, is perfect. They all make mistakes - ours is no exception. However, an intelligent system will make a given mistake only once. This is how:

Eventually the validity of a given signature  $X$  will be contested, by a complaint arriving from the environment of our system. Two situations are possible here:

1. Our system authenticated a forged signature, or
2. A valid signature has been rejected.

In the first situation we will have to insert the forged signature  $X$  into the set of forgeries, and reduce the size of some reference ball(s) accordingly, perhaps deleting some balls that became redundant. If the customer account was opened in the traditional way, then initially the set of forgeries is empty, thus making the system more vulnerable. Nevertheless, it can and will learn, although possibly at higher cost to the banker.

In the second situation, we have to update the customer signature profile by inserting a new ball into it.

In case of authentication challenge, we will use the function `Valid` to determine which of the two situations we are facing. The pseudocode is:

```
(9.9) procedure Fix(X      : in      Signature;
                  C      : in out SigProfile;
                  FSet   : in out ForgedSet)
is
    B      : SigBall;
    mutex  : semaphore := 1;
begin
    if Valid(X, C) then -- we are in situation 1
        parfor i in 1 .. card(C) do
            C.B(i).r := min(C.B(i).r, D(X, C.B(i).S) - 1);
        parend;
        Insert X into Fset;
    else          -- we are in situation 2
        B.S := X; B.r := ∞; -- set initially to infinity
        parfor i in 1 .. card(FSet) do
            radius : integer := D(X, Fset.F(i)) - 1;
            wait(mutex);
            if B.r > radius then B.r := radius; end if;
            signal(mutex);
        parend;
        Insert B into C;
    end if;
    Sort C in descending order of ball radius;
    Remove redundant balls (if any) from C;
end Fix;
```

Having implemented this functionality we get a banking system that continuously learns about customer signatures, is better and better at their authentication, and may even advise some customers that their signatures might be easily forged. It can also forget signatures of customers that

closed their accounts, to keep its adaptation to the work environment. Indeed, it could also add the signatures of its ex-customers to the bank of forgeries, to keep improving its reliability.

## 9. 5. Results

Let us pause for a moment and try to predict what kind of results we should expect. Pure genetic programming systems, neural networks, ant colonies, particle swarms, etc., give good results when exposed to relatively small training and test sets of data, but are laborious to train, and tend to fall apart under ever increasing training and testing workloads [10].

Our banking system is more human-like. Its training is fast and straightforward (it merely learns about new objects by inserting relevant balls into its knowledge bank), but like humans, suffers from the “little knowledge is a dangerous thing” syndrome. Unlike most machine learning algorithms in use today, it is capable of continuous learning, and to continue improving its performance.

Suppose some bank adopts our signature authentication system. When presented the signatures of its first customer, the system knows that it should use these signatures as centers of balls to be inserted into its knowledge bank, but what should be the radii of these balls? Infinity seems plausible, as the system knows of only one customer. But, if infinity were plausible, then one ball to represent that customer would suffice... However, with an infinitely large ball the system would be unable to spot forgeries, even of its single customer. Like any poorly trained human, our system is susceptible to “jumping to conclusions”.

Some caution must then be used here. We ask the customer to offer several sample signatures, as per pseudo-code (9.2) and create several balls of finite size around these signatures. We know, however, that these signatures are unlikely to be the most representative signatures penned by that customer, i.e. we have been overly conservative in determining the ball radii, which should be multiplied by a certain “caution coefficient”  $\alpha$ . To estimate the optimal value of that coefficient, we divided the data set of signatures into a training set of 5 randomly selected signatures, found the value of  $\alpha$  maximizing the overall diagnostic reliability of our system over that training set, and then exposed our system to the customers from the test data in random order.

For handling accounts one account at a time, the results are shown in Table 1. The case of no forgeries used at opening of the account amounts to the use of conventional banking method.

# Signatures on open		System Probabilities		
True	False	P(T1 err.)	P(T2 err.)	Reliability
2	0	0.13542	0.11458	0.75000
2	2	0.14583	0.03125	0.82292
3	0	0.12500	0.10417	0.77083
3	2	0.12500	0.04167	0.83333
4	0	0.13542	0.10417	0.76042
4	2	0.08333	0.05208	0.86458

*Table 1: Overall reliability of signature authentication when handling one customer account at a time. Note that the bank is vulnerable to a financial loss only then committing type 2 error; type 1 error merely annoys the customer.*

Observe that these results are somewhat vulnerable to the choice of customers in our training set, and to the order in which test customers are then presented to the system. Also, a self-respecting bank should keep refining its knowledge bank and fine-tuning its caution coefficient  $\alpha$  while doing business (i.e. performing test transactions). This is what continuous learning is all about.

We tried to increase the system reliability by reducing the dangers stemming from having “too little knowledge” by opening the accounts of all customers first, and only then allowed the bank to do business, by exposing it in random order to transaction of customers from the test set. This simulates the conversion to our system by an existing bank. The ball radii for all customers are typically smaller, because balls belonging to a given customer must be mutually exclusive with balls belonging to other customers, thus making our system overly cautious. The effect of not having the most representative signatures of our customers becomes clearly visible.

The Table 2 shows the results:

# Signatures on open		System Probabilities		
True	False	P(T1 err.)	P(T2 err.)	Reliability
2	0	0.52083	0.0000	0.47917
2	2	0.48958	0.0000	0.51042
3	0	0.52083	0.0000	0.47917
3	2	0.47917	0.0000	0.52803
4	0	0.48958	0.0000	0.51042
4	2	0.48958	0.0000	0.51042

*Table 2: Overall reliability of signature authentication when opening all customer accounts at once. This seems to be the natural way of handling related vision problems, since the environment was already complex (i.e. consisted of many objects) when vision evolved.*

Unfortunately, this attempt was not successful. The overall reliability of the system suffered, although the system did not commit any costly type II errors.

We do not need to convince any banker that it is worthwhile to convert to our system on the training basis of five customers only. Given that humans do not consistently pen their signatures, the system will need more valid samples, acquired while doing business. This inconsistency makes it impossible to reliably verify signatures (humanly or by a machine) on the basis of two, three or four valid samples. That is why we tend to focus on fingerprints, iris scans, and other biometric data when trying to identify humans.

In any case, every cautious banker will be glad to hear that our system is very unlikely to commit costly type II errors. Conversion to our system is possible, but the set of valid signatures of each customer must contain many more samples, taken from the past uncontested transactions.

## 10. Conclusion

We have proposed here a novel methodology for creating intelligent systems capable of

1. Continuous (on the job) learning;
2. Selective forgetting of less useful facts, and so of maintaining its adaptation to an environment that evolves sufficiently slowly;

3. Massive parallelism;
4. Applicability of pattern recognition, identification, clustering and classification outside of the vision domain;
5. Upward and downward scalability to the parallel computer at hand.

In this paper we have focused mainly on machine learning. Forgetting of less useful facts is equally important for real-time systems keen on adaptation to an evolving environment. Banking application was just an example of this. In general, we propose to keep track of usefulness of various balls in the knowledge banks of our systems so as to keep the working set of balls in the systems' "short term memories". The concept of a working set is well known. Details at [8], [9].

As it turns out, it is easier to create reliable but rigid and less intelligent systems with hard-coded knowledge, than adaptive intelligent systems that would learn from their own experience. Such systems will likely need extensive training before being put in the position of responsibility.

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