

Conceptual basis of geostatistics

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Attributes are distributed over space due to a combination of **processes**:

- Process 1: due to **other spatially-distributed attributes**
 - e.g., elevation → temperature; land cover class → vegetation density
- Process 2: due to a **spatial trend** (a function of the coördinates)
 - e.g., distance from source → rock stratum thickness
- Process 3: due to **local effects**
 - e.g., diffusion from a point source → disease/pest incidence in a crop field

- “Auto” = self, i.e., an attribute correlated with itself
- Compare the attribute of one instance with that of another instance of the *same* attribute
- Define how to compare:
 - time: **temporal** autocorrelation (e.g., of time series)
 - space: **spatial** autocorrelation

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$$Z(\mathbf{s}) = Z^*(\mathbf{s}) + \varepsilon(\mathbf{s}) + \varepsilon'(\mathbf{s})$$

(\mathbf{s}) a location in space, designated by a **vector** of coördinates

$Z(\mathbf{s})$ **true** (unknown) value of some property at the location

$Z^*(\mathbf{s})$ **deterministic** component, due to some known or modelled **non-stochastic** process

$\varepsilon(\mathbf{s})$ **spatially-autocorrelated stochastic** component

$\varepsilon'(\mathbf{s})$ pure (“white”) **noise**, no structure

Two types of the deterministic components

$$Z^*(\mathbf{s})$$

- as function of **spatially-distributed covariates** (Process 1)
- as a **trend**: a function of the coördinates (Process 2)
- these can have the same mathematical structure and be determined by the same algorithms
 - covariates: multiple regression, random forests ...
 - trend: low-degree polynomials, generalized additive models, thin-plate splines

How do we fit the universal model?

$Z^*(s)$

- by a **process** model (simulation)
- by an **expert** or heuristic model, e.g., **stratification**, e.g., into map units (polygons)
- by an **empirical-statistical** (“regression”) model in **feature** (“attribute”) space
- by an **empirical-statistical** model in **geographic** space (“trend surface”)

$\varepsilon(s)$

- as a realization of a spatially-correlated **random field** using geostatistics

$\varepsilon'(s)$

- can not not be modelled, but can be quantified → prediction **uncertainty**

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A 2D geographic example

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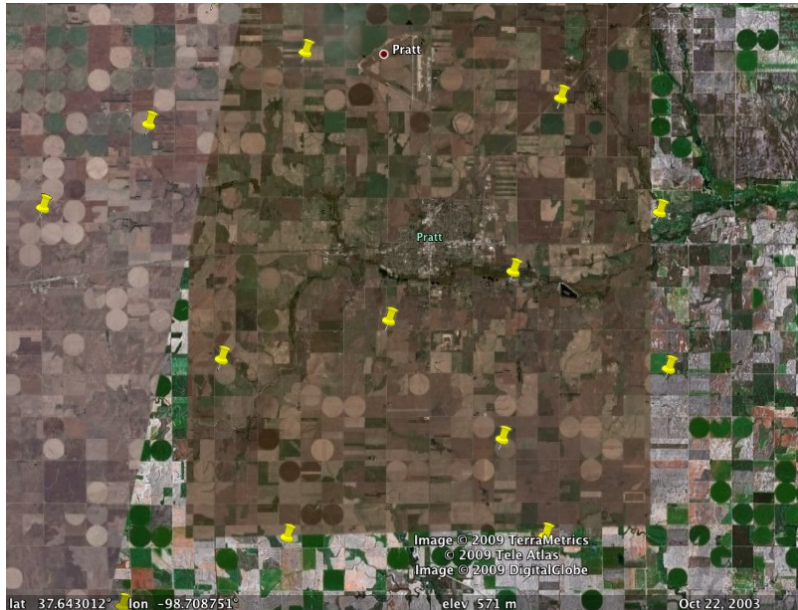
- Source: Olea, R. A., & Davis, J. C. (1999). *Sampling analysis and mapping of water levels in the High Plains aquifer of Kansas* (KGS Open File Report No. 1999-11). Lawrence, Kansas: Kansas Geological Survey.¹
- **attribute**: elevation (US feet) above sea level of the top of an aquifer in Kansas (USA)²; NAVD 88 vertical datum
- **georeference**: observed at a large number of wells, position UTM Zone 14N, NAD83 meters
- Q: **What determines** the spatial variation?
- Q: How can we **model** this from the observations?
- Use the fitted model to **predict** at unsampled locations, either individual locations (proposed new wells) or over a fine-resolution grid

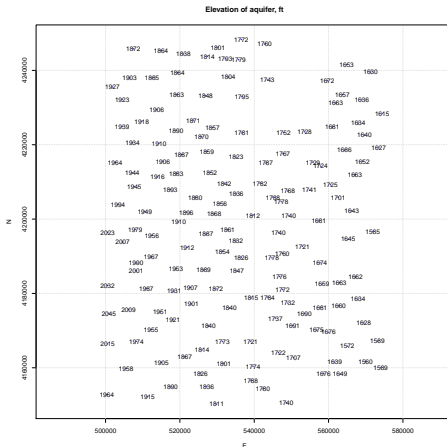
¹Retrieved from http://www.kgs.ku.edu/Hydro/Levels/OFR99_11/

²http://www.kgs.ku.edu/HighPlains/HPA_Atlas/

Reality: $Z(\mathbf{s}) = Z^*(\mathbf{s}) + \varepsilon(\mathbf{s}) + \varepsilon'(\mathbf{s})$

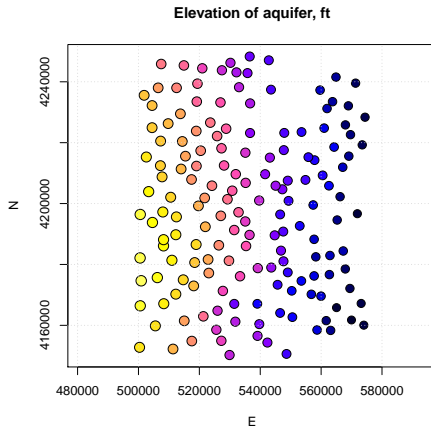
Some well sites on imagery background





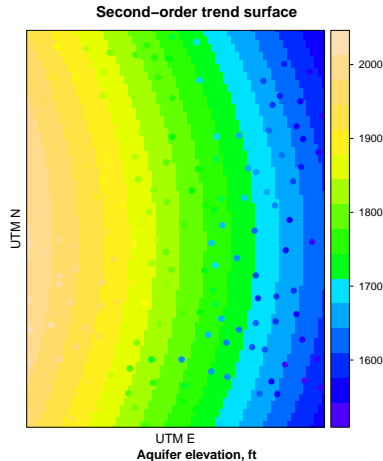
2D georeference, one attribute

Observations – postplot



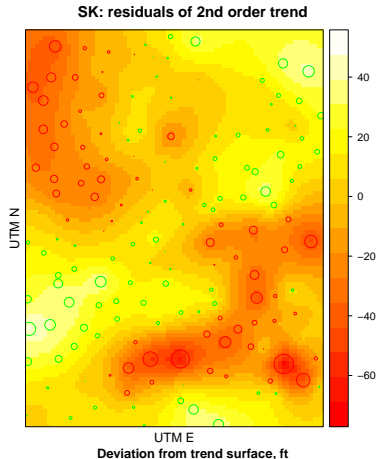
Q: How to divide these observations of $Z(\mathbf{s})$ into $Z^*(\mathbf{s})$, $\varepsilon(\mathbf{s})$, and $\varepsilon'(\mathbf{s})$?

(1) A deterministic trend surface $Z^*(s)$



process: dipping and slightly deformed sandstone rock
modelled with a 2nd-order polynomial (empirical-statistical
model) **trend surface**

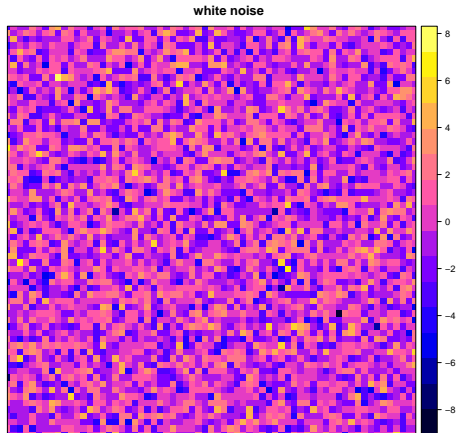
(2) A spatially-correlated random field $\varepsilon(s)$



process: local variations from trend
modelled by variogram modelling of the random field and
simple **kriging**

(3) White noise $\varepsilon'(\mathbf{s})$

We do not know! but **assume** and **hope** it looks like this:



Quantified as **uncertainty** of the other fits

Model with both trend and local variations

$$Z^*(\mathbf{s}) + \varepsilon(\mathbf{s})$$

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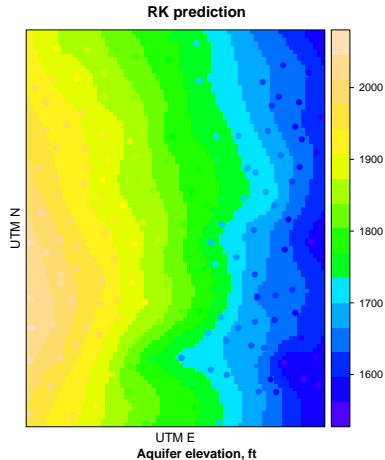
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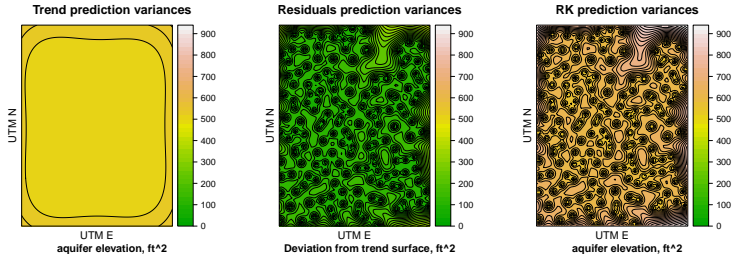
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Computation depends on model form (here: Generalized Least Squares trend + Simple Kriging of GLS residuals)

Model predictions shown on the landscape

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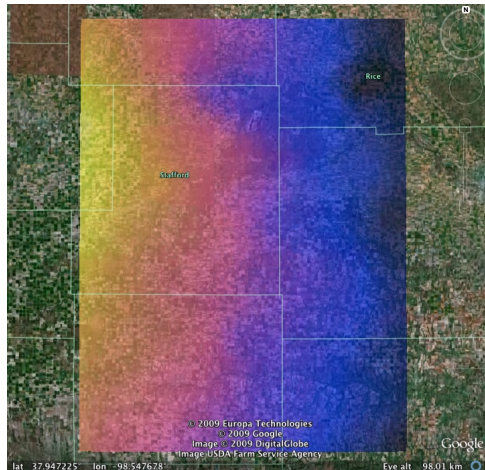
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Google Earth, PNG ground overlay, KML control file

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Example: soil spatial variation (1)

(**s**) area of interest; **discretized** and considered as **blocks** with some finite **support**

$Z(\mathbf{s})$ true block mean and within-block variation

$Z^*(\mathbf{s})$ effect of **soil-forming factors** that can be modelled

- *same* factors → *same* soil properties: Jenny (1941) ‘*clorpt*’ model.
- includes strata (thematic maps units), “continuous” fields
- includes regional geographic trends (e.g., climate gradient)

...

Example: soil spatial variation (2)

...

$\varepsilon(\mathbf{s})$ **spatially-correlated stochastic** component,
modelled in **geographic** space

- local deviations from average effect of soil-forming factors
- some part of this is often **spatially-correlated**; this we can model

...

Example: soil spatial variation (3)

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...

$\varepsilon'(\mathbf{s})$ pure (“white”) **noise**

- non-deterministic and not spatially-correlated
- includes variation at finer scale than support
- includes sampling and measurement imprecision (“error”)

measurement imprecision (all included in “noise”):

- georeferencing / field location
- sampling protocol, sampling procedures
- lab. methods, lab. procedures, lab. quality control

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Conceptual issues with the universal model

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- ① **“Deterministic”** implies that some process always operates the same way with the same inputs.
 - Any deviations are considered noise and included in $\varepsilon(\mathbf{s})$ or $\varepsilon'(\mathbf{s})$.
 - “Deterministic” is **operationally** defined as “we can model it as if it were deterministic”
 - We are *not* really asserting that nature is deterministic.
- ② The **spatially-autocorrelated stochastic** component is assumed to be one realization of a spatially-correlated random process
 - This is usually a convenient fiction to allow modelling.
 - It may include a spatially-correlated deterministic component that we don't know how to model.
 - It is a stochastic process, so there is **uncertainty** which is considered pure noise
- ③ The **“pure noise”** component may also have a structure but at a finer scale than we can measure.
 - It also contains our ignorance about the deterministic process and spatially-correlated random process

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- 4 Predicting from a model of spatial autocorrelation and a set of observations

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- Inferential statistics about a population with **spatial reference**, i.e. **coördinates**:
 - Any number of dimensions (1D, 2D, 3D ...);
 - Any geometry;
 - Any coördinate reference system (CRS), including locally-defined coördinates;
 - There must be defined **distance** and **area** metrics.
- Key point: **observations** and **predictions** of the target variable (and possibly co-variables) are made at **known locations** in geographic space.

Simple case: no deterministic component

- Suppose **no** geographic trend or spatially-distributed covariates
- Then $Z^*(\mathbf{s}) \equiv \mu$, where μ is the **stationary spatial mean**.
- The universal model:

$$Z(\mathbf{s}) = Z^*(\mathbf{s}) + \varepsilon(\mathbf{s}) + \varepsilon'(\mathbf{s}) \quad (1)$$

- becomes:

$$Z(\mathbf{s}) = \mu + \varepsilon(\mathbf{s}) + \varepsilon'(\mathbf{s}) \quad (2)$$

This is a model **assumption!**

- The technical term here is **first-order stationarity**; later we relax this assumption.
- We want to model the **structure** of $\varepsilon(\mathbf{s})$ and ignore the **pure noise** $\varepsilon'(\mathbf{s})$
- the noise sets a lower bound on the precision of predictions made with the fitted model.

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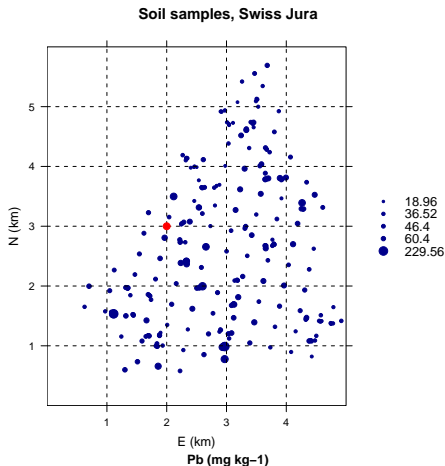
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A “point” observation dataset

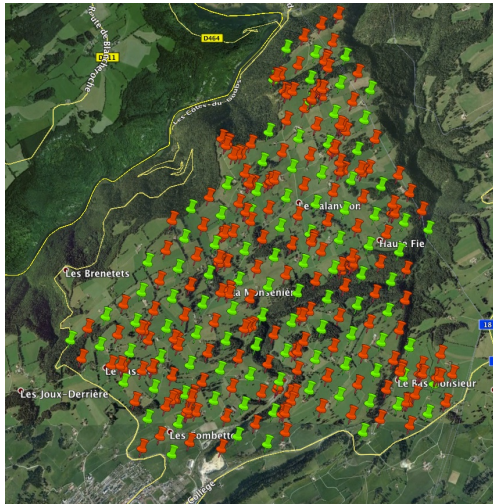
Post plot:

**Symbol size
proportional
to attribute
value**

**Axes are
geographic
coördinate**



Q: Is there spatial autocorrelation of the Pb concentrations?



Detecting local spatial autocorrelation

If there is local spatial autocorrelation, we need to **detect** it (empirically) and then **model** it (mathematically).

- **detection**: h-scatterplot, correlogram or **variogram**
- **modelling**: “authorized” model of spatial covariance

- Any two observations in geographic space are a **point-pair**.
 - We know (1) their coordinate **s**; (2) their **attribute values** (what was measured about them) **z(s)**.
- For an n -observation dataset, there are $(n * (n - 1))/2$ **unique point-pairs**.
 - E.g., 150-point dataset has $150 * 149/2 = 11\ 175$ pairs!
- Each pair is separated in **geographic space** by a **distance** and (if >1D) **direction**.
- Each pair is separated in **feature (attribute) space** by the **difference** between their attribute values.

Semivariance of a point-pair

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- *Define* the **semivariance** γ of one point-pair as:

$$\gamma(\mathbf{s}_i, \mathbf{s}_j) \equiv \frac{1}{2} [z(\mathbf{s}_i) - z(\mathbf{s}_j)]^2$$

- This quantifies the textbfdifference between the **attributes** values at the two points.
- Squared because the order of point-pairs is irrelevant
- 1/2 for technical reasons in the kriging equations (see later)

h -scatterplot: correlation between point-pairs

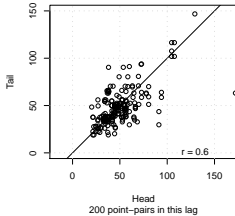
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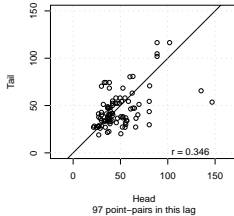
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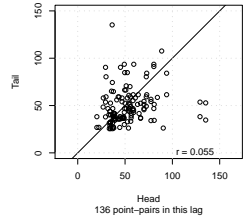
h -scatter plot, lag distance (0, 0.05)



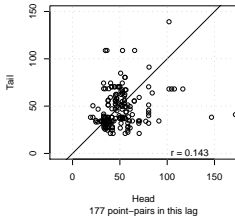
h -scatter plot, lag distance (0.05, 0.1)



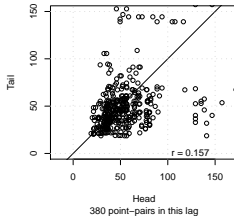
h -scatter plot, lag distance (0.1, 0.15)



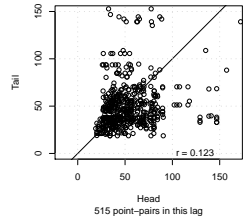
h -scatter plot, lag distance (0.15, 0.2)



h -scatter plot, lag distance (0.2, 0.25)



h -scatter plot, lag distance (0.25, 0.3)



Increasing lag distance $h \rightarrow$ decreasing linear correlation r .

Evidence of spatial autocorrelation from the *h*-scatterplot

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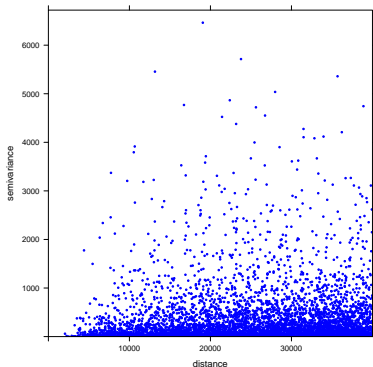
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- Point-pairs compared against the 1:1 line (equal values would be on the line)
- More scatter from the 1:1 line → less linear correlation
- If the **sequence of lags** from close to far also shows **increasing scatter** (i.e., decreasing correlation), this is evidence of local spatial autocorrelation.

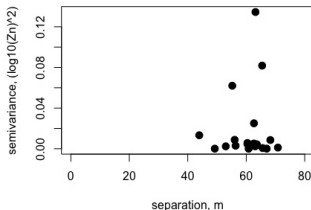
- A scatterplot showing, for *all* point-pairs:

(**x-axis**) the **separation distance** between the two observations

(**y-axis**) their **semivariance** slide



Variogram cloud – detail



```
> (vc <- variogram(logZn ~ 1, meuse,
  cutoff=72, cloud=TRUE))
```

	dist	gamma	left	right
1	70.83784	1.144082e-03	2	1
2	67.00746	9.815006e-05	11	10
3	62.64982	2.504076e-02	22	21
4	53.00000	2.375806e-03	23	22
5	49.24429	8.749351e-05	26	25
6	62.62587	5.128294e-03	33	32
7	65.60488	6.655118e-04	39	38
8	63.07139	2.403081e-03	72	71
9	63.63961	4.318603e-03	76	75
10	60.44005	4.486439e-03	84	9
11	43.93177	1.326441e-02	87	72
12	65.43699	8.178006e-02	87	80
13	56.04463	8.764773e-03	88	73
14	55.22681	6.198261e-02	88	79
15	60.41523	5.680995e-03	123	58
16	60.82763	5.583388e-05	124	52
17	63.15853	1.344946e-01	138	76
18	56.36488	2.996326e-03	139	77
19	68.24222	8.550172e-03	140	91

- Note the **anomalous** point-pair.
- This is difficult to interpret and model, so we summarize this with an **empirical variogram** (see next).

Empirical semivariogram - equation

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Summarize the cloud as **average semivariance** $\bar{\gamma}(\mathbf{h})$ in some **separation range** \mathbf{h}

$$\bar{\gamma}(\mathbf{h}) = \frac{1}{2m(\mathbf{h})} \sum_{i=1}^{m(\mathbf{h})} [z(\mathbf{s}_i) - z(\mathbf{s}_i + \mathbf{h})]^2$$

- $m(\mathbf{h})$ is the number of **point pairs** separated by vector \mathbf{h} , in practice some range of separations (“bin”)
- these are indexed by i
- the notation $z(\mathbf{s}_i + \mathbf{h})$ means the “tail” of point-pair i , i.e., separated from the “head” \mathbf{s}_i by the separation vector \mathbf{h} .

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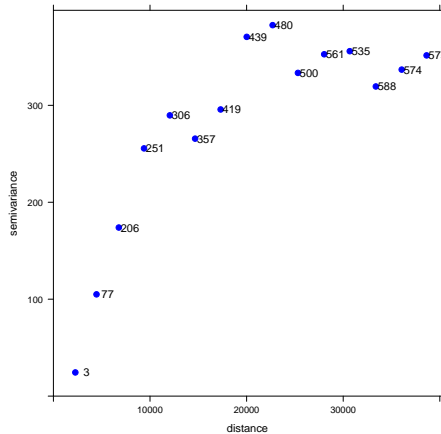
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```
> (v <- variogram(logZn ~ 1, meuse, cutoff=1300, width=90))
      np      dist      gamma
1    41   72.24836 0.02649954
2   212  142.88031 0.03242411
3   320  227.32202 0.04818895
4   371  315.85549 0.06543093
5   423  406.44801 0.08025949
6   458  496.09401 0.09509850
7   455  586.78634 0.10656591
8   466  677.39566 0.10333481
9   503  764.55712 0.11461332
10  480  856.69422 0.12924402
11  468  944.02864 0.12290106
12  460 1033.62277 0.12820318
13  422 1125.63214 0.13206510
14  408 1212.62350 0.11591294
15  173 1280.65364 0.11719960
```

- np = number of point-pairs in bin
- dist = average separation between the point-pairs in bin (here, meters)
- gamma = average semivariance $\overline{\gamma}(\mathbf{h})$ between the point-pairs in bin (here, $\log_{10}\text{Zn}^2$)
- Obvious trend: wider separation \rightarrow larger semivariance

For $>1D$ geometries:

- **distance only**: the **omni-directional** variogram
- **distance and angle**: a **directional** variogram
 - includes a **tolerance angle** and/or maximum width

Is there local spatial autocorrelation?

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- Dependence: a relation between semivariance and separation.
- Closer in **geographic** space means closer in **feature** space.
 - i.e., knowing the attribute value at one observation gives some clue about the value at a “nearby” point
 - The closer to known points, the stronger the clue
- Visualize/infer by **plotting the empirical semivariogram(s)**.
- If there appears to be evidence, then **model**

Terminology of spatial autocorrelation

sill (also **total** Maximum semivariance at any separation

range separation at which the sill is reached or approximated

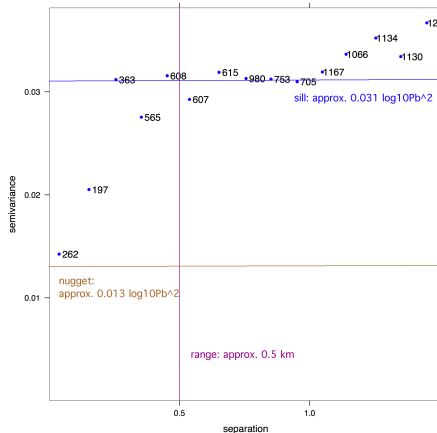
nugget semivariance at zero separation (at a point)

structural sill (also **partial** sill) the total sill less the nugget

- i.e., the portion due to spatial autocorrelation

nugget/sill ratio proportion of total sill due to the nugget, i.e., **unexplainable**

$\log_{10}\text{Pb}$, Jura soil samples



Nugget/sill ratio $\approx 0.42 \rightarrow$ variability not explained $\epsilon'(s)$

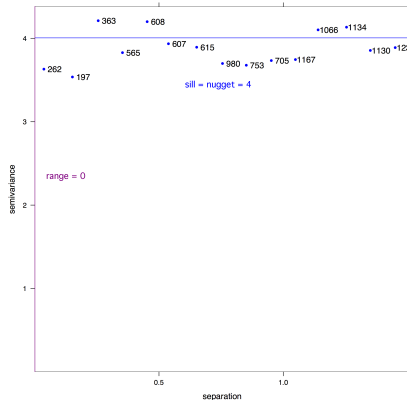
Evidence of spatial autocorrelation from the variogram

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The empirical variogram provides **evidence** that there is **local spatial autocorrelation**.

- The **variation** between point-pairs is **lower** if they are **closer** to each other; i.e. the separation is small.
- There is some distance, the **range** where this effect is noted; beyond the range there is no autocorrelation.
- The **relative magnitude** of the **total sill** and **nugget** give the **strength** of the local spatial autocorrelation; the **nugget** represents completely **unexplained** variation.
- If there is no spatial autocorrelation, we have a **pure nugget** variogram.

Annotated empirical variogram – no spatial autocorrelation



Random fluctuations around sill, due to sampling variation and binning

How reliable is the empirical variogram?

DGR

- Recall: it is based on some **sample** which represents the **population**.
- A different sample of the same size would give a different variogram. **Would they be consistent?**
 - i.e., when modelled (see below) would they result in more-or-less the same model?
- Simulation studies: e.g., Webster, R., & Oliver, M. A. (1992). *Sample adequately to estimate variograms of soil properties*. Journal of Soil Science, 43(1), 177–192.
- Conclusion: **150 to 200** observations allow reliable reconstruction of a known variogram model in the **isotropic** case.

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Modelling spatial autocorrelation

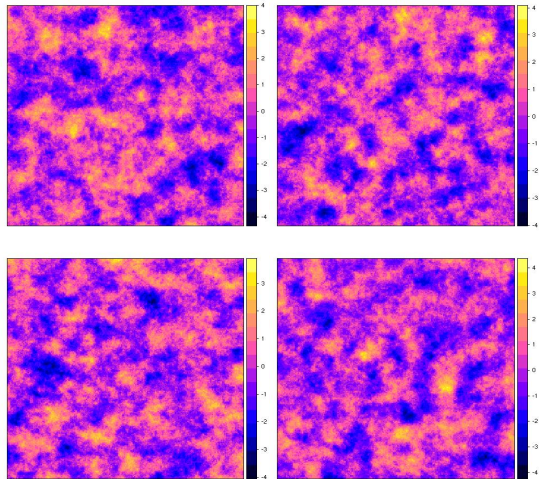
- Aim: To fit a **mathematical model** to an empirical variogram
- This model must be based on some **theory** – this is a modelling **assumption**.
- Theory: **random fields**³

³Source: Webster, R., & Oliver, M. A. (2001). *Geostatistics for environmental scientists*, John Wiley& Sons, Ltd.

Spatially-autocorrelated random processes

- Assumption: The observed attribute values are only **one of many possible realisations** of a **random** (“stochastic”) **process**
- This process is **spatially autocorrelated**, i.e., observations are **not independent**
- The result is called a **random field**
- Different stochastic processes are represented by different **models of spatial covariance**
- There is **only one reality** (which is sampled)
- From our one reality, we need to **infer the process** that produced it
- This dictates the proper **authorized variogram** (or, covariance) **function**.

Four realizations of the same random field



256 x 256 grid; Spherical model; range 25; no nugget

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- 1 **Assume** reality is one realization of a **regionalized variable** (structure to be determined)
- 2 **Assume** any spatial autocorrelation has the **same structure** everywhere
 - This is **2nd-order stationarity**
- 3 Make observations; summarize as an **empirical variogram**
- 4 Select a **model of spatial autocorrelation**
- 5 **Parameterize** (fit) the selected model to the empirical variogram

Selecting a model of spatial covariance

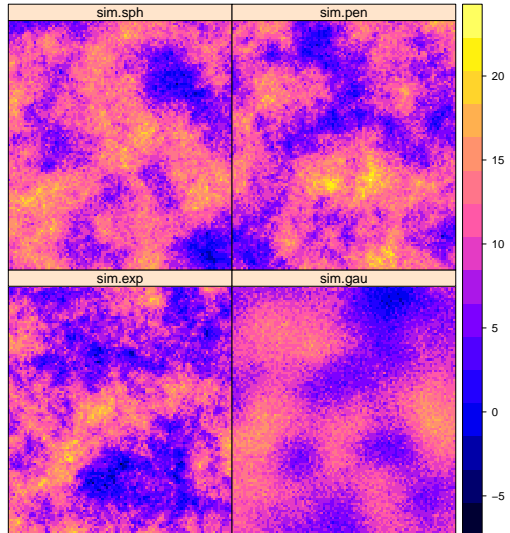
Various methods, more-or-less in order of preference:

- ① What is known about the **spatial process** that produced the field
- ② **Previous studies** of the same variable in similar circumstances
- ③ Visual assessment of the variogram form
- ④ Try to fit many, automatic selection by “best” fit
- ⑤ Problem with “best” fit: depends on:
 - ① variogram cutoff, bin width
 - ② criterion for “best”, e.g., more weight to more point-pairs and closer separations
 - ③ other forms may fit almost as well

Four regionalized covariance models

same
model
parameters

Different
processes



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Variogram model equations (1)

- Only some forms are **authorized**, i.e., will lead to positive-definite kriging matrices (see below). We review a few common models.
- All can be raised by the **nugget** variance c_0 .
- **Exponential** model: sill c , **effective range** $3a$

$$\gamma(h) = c \left(1 - e^{\left(-\frac{h}{a}\right)} \right)$$

- Autocorrelation decreases exponentially with separation – the minimum spatial dependence.
- This is an *asymptotic* model: variance **approaches** a sill at some **effective range**, by convention, where $\gamma = 0.95c$.

Variogram model equations (2)

Gaussian model: sill c , effective range $\sqrt{3}a$:

$$\gamma(h) = c \left(1 - e^{-\left(\frac{h}{a}\right)^2} \right)$$

This has strong spatial continuity near the origin
(0-separation), e.g., water table elevation, smoothly-varying
terrain properties

Variogram model equations (3)

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Matérn model family: *generalizes* the Exponential, Power, Logarithmic and Gaussian models

$$\gamma(h) = c \left(1 - \frac{1}{2^{\kappa-1} \Gamma(\kappa)} \left(\frac{h}{a} \right)^{\kappa} K_{\kappa} \left(\frac{h}{a} \right) \right)$$

- smoothness parameter is κ ; this adjust the variogram model to the process.
- small κ implies that the spatial process is rough, large κ smooth.
- K_{κ} is a modified Bessel function of the second kind
- Γ is the Gamma function (generalization of the factorial function)
- if $\kappa = 0.5$ this reducesto the exponential model
- if $\kappa = \infty$ this reduces to the Gaussian model
- most common values are $\kappa = 0.5, 1, 1.5, 2$

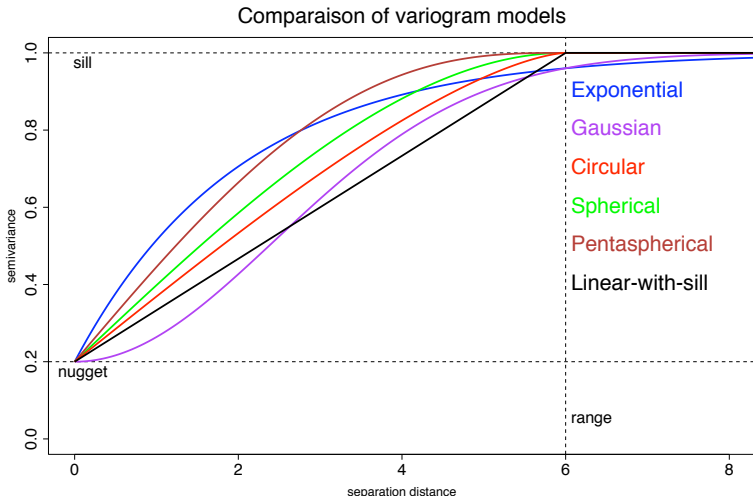
Spherical model: sill c , range a

$$\gamma(h) = \begin{cases} c \left(\frac{3}{2} \frac{h}{a} - \frac{1}{2} \left(\frac{h}{a} \right)^3 \right) & : h < a \\ c & : h \geq a \end{cases}$$

This is linear near the origin, reaches the sill c at the range a and is then constant, with a “shoulder” transition between.

It is often applied when the variable occurs in somewhat homogeneous **patches** with gradual boundaries, e.g., vegetation density, soil properties.

Comparing variogram models – same parameters



Fitted variogram models to same empirical variogram

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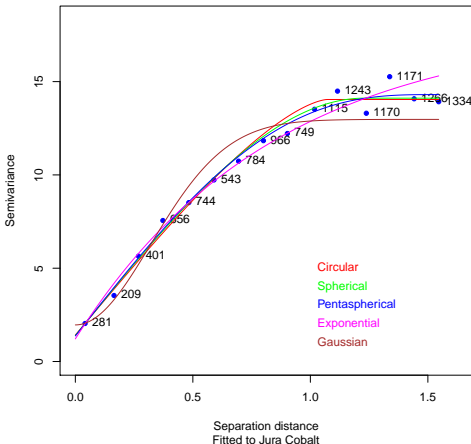
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Variogram model forms



- Any **linear combination** of authorized models is also authorized
- Models with > 1 spatial structure at **different ranges**
- Common example: **nugget + structural**
- e.g. nugget + exponential

$$\gamma(h) = c_0 + c_1 \left(1 - e^{\left(-\frac{h}{a}\right)} \right)$$

- Structure at two ranges: e.g., **nugget + exponential + exponential**

$$\gamma(h) = c_0 + c_1 \left(1 - e^{\left(-\frac{h}{a_1}\right)} \right) + c_2 \left(1 - e^{\left(-\frac{h}{a_2}\right)} \right)$$

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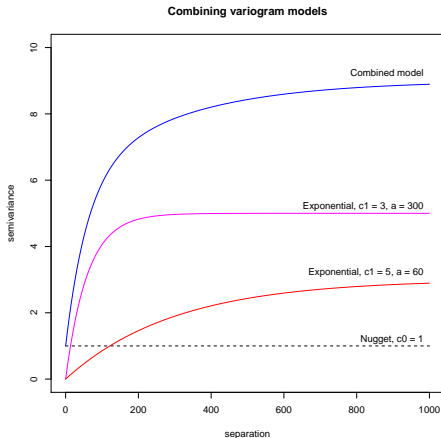
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Models nugget, short range ($3a = 180$) and long range ($3a = 900$) structures

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Predicting from a model of spatial autocorrelation and a set of observations

- Once we have a variogram model, it can be used to **predict** at unobserved locations.
- Model without trend: $Z(\mathbf{s}) = \mu + \varepsilon(\mathbf{s}) + \varepsilon'(\mathbf{s})$
- The realization of the random field at point \mathbf{s} is:
 - some **mean value** μ ; plus ...
 - ... a **spatially-autocorrelated** random component $\varepsilon(\mathbf{s})$, with a defined covariance structure (e.g., a variogram model); plus ...
 - ... pure **noise** $\varepsilon'(\mathbf{s})$: nugget and lack of spatial correlation with increasing separation
 - Both the expected value (1st-order) and covariance structure (2nd-order) are **stationary**: the same everywhere in the field

All of these have **no theory of spatial autocorrelation**, they have *ad hoc* implicit models of spatial structure:

- nearest neighbour (Thiessen polygons, Voronoi tessellation of space)
- average of nearest k -neighbours
- average of nearest k -neighbours weighted by inverse distance to some power
- average of all neighbours within some radius
- average of all neighbours within some radius weighted by inverse distance to some power
- ... with de-clustering of compact groups of known points

Choice of k , radius by cross-validation.

A geostatistical prediction method: Ordinary Kriging (OK)

- The estimated value \hat{z} at a point \mathbf{x}_0 is predicted as the **weighted average** of the values at *all* sample points \mathbf{x}_i :

$$\hat{z}(\mathbf{x}_0) = \sum_{i=1}^N \lambda_i z(\mathbf{x}_i)$$

- The weights λ_i assigned to the sample points **sum to 1**: $\sum_{i=1}^N \lambda_i = 1$, therefore, the prediction is **unbiased**.
- Many other interpolators (e.g., inverse distance) are also linear unbiased, but OK is the “**best**” of all possible weightings

In what sense is OK the “best” predictor?

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- OK is called the “**Best Linear Unbiased Predictor**” (BLUP)
- “best” \equiv lowest **prediction variance** of all possible weightings
 - i.e., each prediction has the smallest possible confidence interval
- This criterion is used to derive the **OK system of equations**, which is solved to determine the **weights** for each sample point
- Weights depend on the **spatial covariance structure** as modelled by the **variogram model**.
- Spatial structure **between observations**, as well as **between observations and a prediction point**, is accounted for.

- The **prediction** and its variance are only as good as the **model of spatial structure**.
- Points **closer** to the point to be predicted have **larger weights**, according to the modelled **spatial dependence**
- **Clusters** of points “**reduce to**” single equivalent points
 - i.e., over-sampling in a small area can't bias result
 - automatically de-clusters
- Closer sample points “**mask**” further ones in the same direction
- Error estimate is based only on the **spatial configuration of the sample**, not the data values

Experimenting with OK: E{Z}-Kriging

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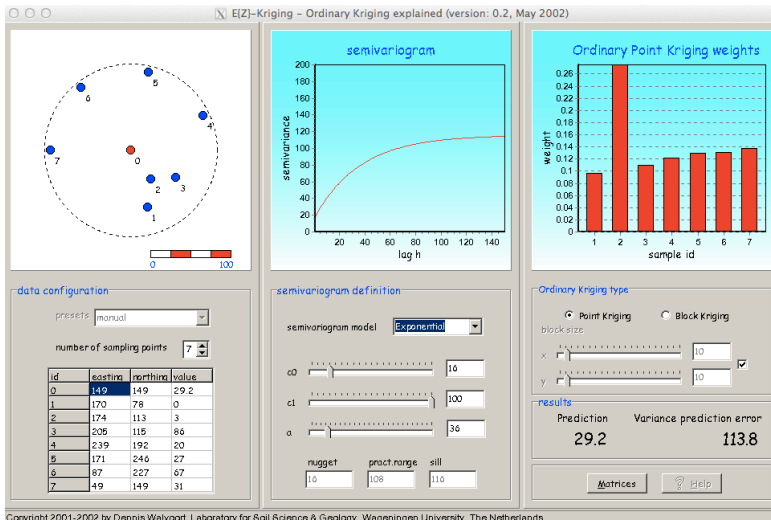
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https://wiki.52north.org/AI_GEOSTATS/SWEZKriging

Derivation of the OK system of equations

- Aim: **minimize the prediction variance**, subject to the **unbiasedness** and **spatial covariance** constraints.
- Two ways to derive the OK system:
 - Regression** As a special case of **weighted least-squares** prediction in the **generalized linear model** with orthogonal projections in linear algebra
 - Minimization** **Minimizing the kriging prediction variance** with calculus
- Approach (1) is mathematically more elegant and is an extension of linear modelling theory.
- Approach (2) is an application of standard minimization methods from differential calculus; but is not so transparent, because of the use of LaGrange multipliers.

Matrix form of the Ordinary Kriging system

$$\mathbf{A}\lambda = \mathbf{b}$$

$$\mathbf{A} = \begin{bmatrix} \gamma(\mathbf{x}_1, \mathbf{x}_1) & \gamma(\mathbf{x}_1, \mathbf{x}_2) & \cdots & \gamma(\mathbf{x}_1, \mathbf{x}_N) & 1 \\ \gamma(\mathbf{x}_2, \mathbf{x}_1) & \gamma(\mathbf{x}_2, \mathbf{x}_2) & \cdots & \gamma(\mathbf{x}_2, \mathbf{x}_N) & 1 \\ \vdots & \vdots & \cdots & \vdots & \vdots \\ \gamma(\mathbf{x}_N, \mathbf{x}_1) & \gamma(\mathbf{x}_N, \mathbf{x}_2) & \cdots & \gamma(\mathbf{x}_N, \mathbf{x}_N) & 1 \\ 1 & 1 & \cdots & 1 & 0 \end{bmatrix}$$

$$\lambda = \begin{bmatrix} \lambda_1 \\ \lambda_2 \\ \vdots \\ \lambda_N \\ \psi \end{bmatrix} \quad \mathbf{b} = \begin{bmatrix} \gamma(\mathbf{x}_1, \mathbf{x}_0) \\ \gamma(\mathbf{x}_2, \mathbf{x}_0) \\ \vdots \\ \gamma(\mathbf{x}_N, \mathbf{x}_0) \\ 1 \end{bmatrix}$$

- **kriging weights** λ_i to be assigned to each observation point
- **semivariances** γ between
 - 1 point to be predicted \mathbf{x}_0 and observation points \mathbf{x}_i ;
 - 2 pairs of observation points $(\mathbf{x}_i, \mathbf{x}_j)$
- **LaGrange multiplier** ψ which enters in the prediction variance

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- This is a system of $N + 1$ equations in $N + 1$ unknowns, so can be solved uniquely for the weights vector λ .

$$\lambda = \mathbf{A}^{-1} \mathbf{b}$$

- But to compute the matrix inverse \mathbf{A}^{-1} the \mathbf{A} matrix (spatial structure) must be **positive definite**
- This is guaranteed for **authorized models** of spatial covariance

- Now we can **predict** at the point, as a weighted sum:

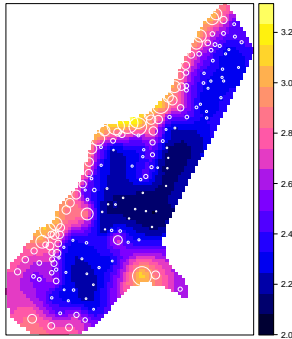
$$\hat{Z}(\mathbf{x}_0) = \sum_{i=1}^N \lambda_i z(\mathbf{x}_i)$$

- The **kriging variance** at a point is computed as:

$$\hat{\sigma}^2(\mathbf{x}_0) = \mathbf{b}^T \boldsymbol{\lambda}$$

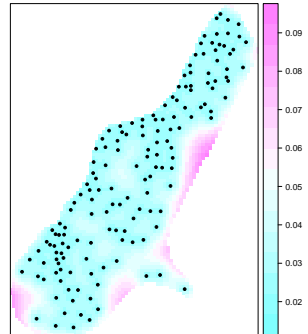
Ordinary kriging (OK) predictions and variances

OK prediction, log-ppm Zn



Predictions, log(Cd)
With postplot

OK prediction variance, log-ppm Zn²



Variance, Meuse log(Cd)²
With sample points

Characteristics of OK prediction

- 1 **smooth**: moving across the map, the kriging weights change smoothly, because the distance changes smoothly
- 2 Prediction is “best” (given the model and data) at each point separately
- 3 But the map is not realistic as a whole (smoother than reality)
- 4 Pure noise at each point represented by the prediction variance
- 5 Variance depends on the **configuration of the sample points**, *not* the **data values**!

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Geostatistics with the universal model

- Recall: the universal model is: $Z(\mathbf{s}) = Z^*(\mathbf{s}) + \varepsilon(\mathbf{s}) + \varepsilon'(\mathbf{s})$
- In the previous section we replaced $Z^*(\mathbf{s})$ with a constant $\mu \rightarrow$ **1storder stationarity**.
- Now we return to the full model: both the **deterministic** and **spatially-autocorrelated** must be modelled
- Question: How to separate the effects? or how to model them in one step?

Universal Kriging (UK)

- This is a **mixed predictor** which includes a **global trend** as a function of the **geographic coördinates** in the kriging system, as well as **local spatial dependence**.
- **Example:** The depth to the top of a given sedimentary layer may have a regional trend, expressed by geologists as the **dip** (angle) and **strike** (azimuth). However, the layer may also be **locally** thicker or thinner, or deformed, with spatial autocorrelation in this local structure – the **residuals** of the trend surface.
- UK is recommended when there is evidence of **1st-order non-stationarity**, i.e. the **expected value** varies across the map, but there is still **2nd-order stationarity** of the **residuals** from this trend.

- The trend is modelled as a **linear combination** of p **base functions** $f_j(s)$ and p unknown constants β_j (these are the **parameters** of the base functions):

$$Z^*(s) = \sum_{j=1}^p \beta_j f_j(s)$$

- Base functions for **linear** drift:

$$f_0(s) = 1, f_1(s) = x_1, f_2(s) = x_2$$

where s_1 is one coördinate (say, E) and s_2 the other (say, N)

- Note that $f_0(s) = 1$ estimates the global mean (as in OK).
- Base functions for **quadratic** drift: also include second-order terms:

$$f_3(s) = s_1^2, f_4(s) = s_1 s_2, f_5(s) = s_2^2$$

- The **unbiasedness** condition is expressed with respect to the **trend** as well as the overall mean (as in OK):

$$\sum_{i=1}^N \lambda_i f_k(\mathbf{s}_i) = f_k(\mathbf{s}_0), \quad \forall k$$

- The expected value at each point of all the **functions** must be that predicted by that function. The first of these is the overall mean (as in OK).
- Example** for a linear trend: If $f_1(\mathbf{s}_0) = s_1$, then at each point \mathbf{s}_0 the expected value must be s_1 , i.e. the point's E coördinate:

$$\sum_{i=1}^N \lambda_i s_i = s_1$$

This is a **further restriction** on the weights λ .

$$\mathbf{A}_U \lambda_U = \mathbf{b}_U$$

$$\mathbf{A}_U = \begin{bmatrix} \gamma(\mathbf{x}_1, \mathbf{x}_1) & \cdots & \gamma(\mathbf{x}_1, \mathbf{x}_N) & 1 & f_1(\mathbf{x}_1) & \cdots & f_k(\mathbf{x}_1) \\ \vdots & \cdots & \vdots & \vdots & \vdots & \cdots & \vdots \\ \gamma(\mathbf{x}_N, \mathbf{x}_1) & \cdots & \gamma(\mathbf{x}_N, \mathbf{x}_N) & 1 & f_1(\mathbf{x}_N) & \cdots & f_k(\mathbf{x}_N) \\ 1 & \cdots & 1 & 0 & 0 & \cdots & 0 \\ f_1(\mathbf{x}_1) & \cdots & f_1(\mathbf{x}_N) & 0 & 0 & \cdots & 0 \\ \vdots & \vdots & \vdots & \vdots & \vdots & \vdots & \vdots \\ f_k(\mathbf{x}_1) & \cdots & f_k(\mathbf{x}_N) & 0 & 0 & \cdots & 0 \end{bmatrix}$$

The upper-left $N \times N$ block is the spatial correlation structure (as in OK)

The lower-left $k \times n$ block and its transpose in the upper-right are the trend predictor values at sample points

The rest of the matrix fits with λ_U and \mathbf{b}_U to set up the solution.

$$\lambda_{\mathbf{U}} = \begin{bmatrix} \lambda_1 \\ \dots \\ \lambda_N \\ \psi_0 \\ \psi_1 \\ \dots \\ \psi_k \end{bmatrix} \quad \mathbf{b}_{\mathbf{U}} = \begin{bmatrix} y(\mathbf{x}_1, \mathbf{x}_0) \\ \vdots \\ y(\mathbf{x}_N, \mathbf{x}_0) \\ 1 \\ f_1(\mathbf{x}_0) \\ \vdots \\ f_k(\mathbf{x}_0) \end{bmatrix}$$

The $\lambda_{\mathbf{U}}$ vector contains the N weights for the sample points and the $k + 1$ LaGrange multipliers (1 for the overall mean and k for the trend model)

$\mathbf{b}_{\mathbf{U}}$ is structured like an additional column of $\mathbf{A}_{\mathbf{U}}$, but referring to the point to be predicted.

- Same as OK: a weighted linear combination of values at known points:

$$\hat{Z}(\mathbf{x}_0) = \sum_{i=1}^N \lambda_i z(\mathbf{x}_i)$$

- But, **the weights** λ_i for each sample point take into account both the **global trend** and **local spatial autocorrelation of the trend residuals**.
- The UK system must include both of these.

Computing the empirical semivariogram for UK

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- The semivariances γ are based on the **residuals**, not the original data, because the **random field** part of the spatial structure applies only **after** any trend has been removed.
- How to obtain?
 - 1 Calculate the **best-fit surface**, with the same base functions to be used in UK;
 - 2 **Subtract** the trend surface at the data points from the data value to get residuals;
 - 3 Compute the **variogram** of the **residuals**.
 - 4 Note that `gstat::variogram` can do this in one step.
- **Problem:** the trend should have taken the spatial correlation into account!

Characteristics of the residual variogram

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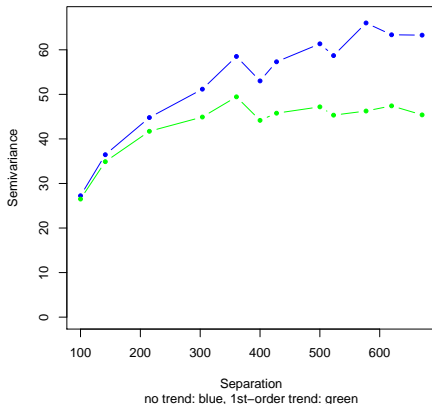
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- If there is a strong trend, the variogram model **parameters** for the residuals will be very different from the original variogram model, since **the global trend has taken out some of the variation**, i.e. that due to the long-range structure.
- The usual case is:
 - lower **sill** (less total variability)
 - shorter **range** (long-range structure removed)
- In theory, the **nugget** should be unchanged (residual variance at a point is not removed by a trend)

Example original vs. residual variogram

Variograms, Oxford soils, CEC (cmol+ kg⁻¹ soil)



Note lower (partial, total) sill, shorter range, same nugger

Universal Kriging: Local vs. Global trends

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As with OK, UK can be used two ways:

- **Globally:** using **all** sample points when predicting each point
- **Locally**, or in **patches**: restricting the sample points used for prediction to some **search radius** (or sometimes **number of neighbours**) around the point to be predicted

Why use UK in a neighbourhood?

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- This allows the **trend surface to vary** over the study area, since it is **re-computed at each prediction point**
- Appropriate to smooth away some **local variation in a trend**
- Difficult to justify theoretically
- Note that the **residual variogram** was not computed in patches, but assuming a global trend
- Leads to some patchiness in the map
- There should be some **evidence of patch size**, perhaps from the original (*not* residual) variogram; this can be used as the search radius.

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Kriging with External Drift (KED)

DGR

A universal model of spatial variation

Example – aquifer elevation

Example – soil spatial variation

Conceptual issues

Geostatistics

Definition

Detecting spatial autocorrelation

Modelling spatial autocorrelation

Variogram models

Prediction

Universal Kriging

Kriging with External Drift (KED)

- This is a mixed interpolator that includes **feature-space predictors**, rather than geographic coördinates (as in UK).
- The **mathematics are exactly as for UK**, but the *base functions* are different.
- UK vs. KED:
 - In UK, the base functions refer to the **grid coördinates**; these are by definition known at any prediction point.
 - In KED, the base functions refer to some **feature-space covariates** ...
 - ... measured at the sample points (so we can use it to set up the predictive equations) and
 - **also known at all prediction points** (so we can use it in the prediction itself).

There are two kinds of feature-space covariates:

- ① **strata**, i.e., factors, categorical variables. Examples: soil type, flood frequency class
 - Base function: $f_k(\mathbf{s}) = 1$ iff sample or prediction point \mathbf{s} is in class k , otherwise 0 (class indicator variable)
- ② **continuous covariates**. Examples: elevation, NDVI
 - Base function: $f_k(\mathbf{s}) = v(\mathbf{s})$, i.e. the value of the predictor at the point.

Note that $f_0(\mathbf{s}) = 1$ for all models; this estimates the global mean (as in OK).