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MIXTURE OF MULTIVARIATE GAUSSIAN PROCESSES FOR CLASSIFICATION OF IRREGULARLY SAMPLED SATELLITE IMAGE TIME-SERIES

BY ALEXANDRE CONSTANTIN^{1,*}, MATHIEU FAUVEL² AND STÉPHANE GIRARD^{1,†}

¹Université Grenoble Alpes, Inria, CNRS, Grenoble INP, LJK, 38000 Grenoble, France, *alexandre.constantin@inria.fr; [†]stephane.girard@inria.fr

²CESBIO, Université de Toulouse, CNES/CNRS/INRAe/IRD/UPS, 31000 Toulouse, France, mathieu.fauvel@inrae.fr

The classification of irregularly sampled Satellite image time-series (SITS) is investigated in this paper. A multivariate Gaussian process mixture model is proposed to address the irregular sampling and the multivariate nature of the time-series. The spectral and temporal correlation is handled using a Kronecker structure on the covariance operator of the Gaussian process. The multivariate Gaussian process mixture model allows both for the classification of time-series and the imputation of missing values. Experimental results on simulated and real SITS data illustrate the importance of taking into account the spectral correlation to ensure a good behavior in terms of classification accuracy and reconstruction errors.

1. Introduction. Satellite images availability has exponentially grown in the last decade. Thanks to free data access policy, optical satellite image time-series (SITS) such as *Landsat* or *Sentinel-2*, offer an unique opportunity to monitor the state and evolution of our living planet. Therefore, SITS have found many applications in ecological monitoring (Li et al., 2018; Fauvel et al., 2020), meteorology (Liu, Gopal and Kalagnanam, 2018; Bertolacci et al., 2019) or agricultural system mapping (Useya and Chen, 2018; Moeini Rad et al., 2019; Feng et al., 2019), among others.

SITS are characterized by their spatial and spectral resolutions, and their revisit cycle. The spatial resolution corresponds to the size of a pixel on the ground, *e.g.*, a square of 10 meters while the spectral resolution is related to the number of wavelengths collected by the sensor, ranging typically in the visible and near infra-red part of the spectrum (Manolakis, Lockwood and Cooley, 2016). The revisit cycle stands for the time between two acquisitions over the same location: SITS have constant and short (*e.g.* few days) revisit time. Hence, for a given temporal period, a pixel is the collection of spectral measurements made at different times over the same location.

These properties lead to an unprecedented amount of numerical data, for which statistical methods are used to extract meaningful information such as land cover, crops yields ... For a pixel-wise based analysis, the predictor variables are multivariate time-series and the output variables represent the information to be extracted. While spatial independence is usually assumed (Landgrebe, 2005), temporal and spectral correlations are commonly taken into account in statistical models (Lopes et al., 2017).

However, external random meteorological factors interfere with the availability of the acquired data at the pixel scale. Indeed, as displayed in Fig. 1, shadows and clouds result in missing data in the time-series. Furthermore, orbital trajectory generates an irregular temporal sampling: Even though the acquisition scheme is regular, acquisition days are different for pixels located at different places (Inglada et al., 2017). As such, each pixel of the SITS

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Fig 1. True color Sentinel-2 satellite image time-series. Data were acquired in 2018 at different time steps over the area of Toulouse, France (images were downloaded from Theia Land Data Center: http://www.theia-land.fr/en/presentation/products).



Fig 2. Illustration of the irregular temporal sampling for the SITS used in this work. Three time-series at different locations for one spectral band are reported: A black dot indicates that the pixel is clear (no shadow or cloud) at the considered time, and a light-gray dot indicates that the pixel has been tagged as clouds or shadows by the data provider.

has its own size in the temporal domain: Fig. 2 illustrates the irregular temporal sampling on the data under consideration in this paper.

Specific models are thus required to properly analyze such time-series, as described in Section 2. Conventional approaches usually start by resampling the data onto a common temporal grid. In this work, we aim at analyzing irregularly sampled multidimensional SITS without any temporal resampling. In particular, the supervised pixel classification task is considered, *i.e.* the assignment of each pixel of the time-series to a predefined class.

To this end, a mixture of multivariate Gaussian Processes is proposed. A linear dependence model is assumed between the spectral variables leading to a separable covariance function in time and spectral domains. The resulting model provides statistical information on the underlying process for each class (mean and covariance functions) and scales linearly w.r.t. the number of samples.

The remainder of this paper is organized as follows. Section 2 reviews the state-of-the-art on classification with missing data and Gaussian processes. The statistical model is introduced in Section 3 while inference aspects are discussed in Section 4 including the estimation of the model parameters, the supervised classification, and the imputation of missing values. These statistical procedures are validated on simulated data in Section 5. Section 6 is dedicated to the application of our methodology to the classification of SITS from a Sentinel-2 data-set. Section 7 concludes with a discussion on possible extensions of this work.

2. Related Work. This section briefly reviews state-of-the-art methods for model-based classification, classification dealing with missing values and classification with Gaussian processes.

2.1. Supervised model-based classification. Supervised model-based classification (also referred to as model-based discriminant analysis) starts from a training set of *n* independent realizations from a random pair $(\mathbf{Y}, Z) \in E \times \{1, ..., C\}$ and assumes that the conditional distribution of $\mathbf{y}|Z = c$ belongs to some parametric family: $p(\mathbf{y}|Z = c) = p_c(\mathbf{y}; \boldsymbol{\theta}_c)$, for all $c \in \{1, ..., C\}$ and $\mathbf{y} \in E$, where *E* is an arbitrary space. Letting $\pi_c = \mathbb{P}(Z = c)$, the marginal distribution of \mathbf{Y} can be written as a finite mixture

$$p(\mathbf{y}) = \sum_{c=1}^{C} \pi_c p_c(\mathbf{y}; \boldsymbol{\theta}_c),$$

whose parameters can be estimated by the maximum likelihood principle. A non-labeled observation can then be classified thanks to the Maximum a posteriori (MAP) criteria:

$$\hat{c} = \underset{c \in \{1, \dots, C\}}{\operatorname{arg max}} p(Z = c | \mathbf{y}) = \underset{c \in \{1, \dots, C\}}{\operatorname{arg max}} \pi_c p_c(\mathbf{y}; \boldsymbol{\theta}_c),$$

thanks to Bayes' rule. When $E = \mathbb{R}^q$, the multivariate Gaussian distribution is often adopted for $p_c(\mathbf{y}; \boldsymbol{\theta}_c)$ and gives rise to the well-known Quadratic discriminant analysis (QDA) method. We refer to (Hastie, Tibshirani and Friedman, 2009, Section 4.3) for a discussion on the advantages and drawbacks of QDA and for possible extensions. Recent studies extend the model-based classification framework to non-Gaussian distributions such as the skew-normal distributions (Theodossiou, 1998; Chamroukhi, 2017) to deal with asymmetric data, or *t*distributions (Andrews and McNicholas, 2012; Murray, Browne and McNicholas, 2017) to deal with outliers. We refer to (Bouveyron et al., 2019, Chapter 9) for an in-depth review. The case $E = \mathbb{R}^q$ also encompasses the situation of discretized time-series on a common grid. Specific models can be then defined, as in (Povinelli et al., 2004) for temporal signatures.

If *E* is discrete, including for example the case of categorical data, extensions focus on the multinomial (Celeux and Govaert, 1991) or the Dirichlet (Bouguila, Ziou and Vaillancourt, 2003) distributions. In the case of ordinal data, other extensions are proposed using a dedicated model of the process generating data (Biernacki and Jacques, 2016). Finally, when *E* is more complex, *e.g.* infinite dimensional, non-parametric techniques are used. Kernel methods are probably the most popular non-parametric techniques in this situation (Hofmann, Schölkopf and Smola, 2008). Recall that a kernel is a positive definite function that corresponds to a dot product in a feature space. It allows for the construction of non-linear and non-parametric classifiers on *E* without computing explicitly the feature space. Kernels can be defined, for instance, on strings (Lodhi et al., 2002), graphs (Kriege, Johansson and Morris, 2020), vector-valued functions (Alvarez, Rosasco and Lawrence, 2012; Flaxman et al., 2019), or combinations of several data types (Bouveyron, Fauvel and Girard, 2015).

2.2. Classification with missing data. When dealing with remote sensing data, *i.e.* spatial-spectro temporal data such as SITS, handling missing values (Shen et al., 2015) is a recurrent problem. Classification dealing with missing data occurs when some inputs in the training set are incomplete, *i.e.* the number of available coordinates in y can be different from one sample to another, see (Schafer, 1997; García-Laencina, Sancho-Gómez and Figueiras-Vidal, 2010; Bagnall et al., 2017) for reviews.

Three main approaches can be found in the literature. A first solution is to impute missing values before the classification itself. The pre-processing gives rise to a training set with observations re-sampled on a common grid that can be considered as vectors in a finite space $E = \mathbb{R}^q$, opening the door to classical model-based classification methods. We refer to (Lin and Tsai, 2020) for a review on imputation techniques. For SITS, Inglada et al. (2017) used such two-stage approaches on Sentinel-2 SITS where a linear interpolation was applied before performing the classification with a Random Forests classifier. Yet, by applying imputation techniques without any connection to the actual processing, propagated errors from the interpolation may degrade the results.

Alternative solutions are based on functional data analysis (Ramsay and Silverman, 2005). Each observation is interpreted as a sample from a random function. As such, it can be approximated by an expansion on some basis functions. The statistical analysis is then performed on the random vectors of coefficients, see (Schmutz et al., 2020) for an application to clustering. Nonparametric smoothing techniques may also adopted, see (Ferraty and Vieu, 2006, Chapter 8) for an overview.

Finally, purely non-parametric methods can also be implemented by defining an appropriate dissimilarity measure between samples of varying size. In the context of time-series, Dynamic time warping (DTW) (Chouakria and Nagabhushan, 2007) is one of the most popular algorithms. It computes an optimal match between two vectors with different lengths. This map defines a dissimilarity that can be used for comparison in order to cluster samples into multiple groups.

2.3. Classification with Gaussian processes. A recent approach for supervised classification is based on the use of Gaussian processes (GPs) in a Bayesian framework. More specifically, Gaussian processes are used as prior distributions on the regression function linking the label Z to the explanatory variable **X**. In the binary classification case, the conditional Bernoulli distribution of Z is defined through a logit transformation: $logit(p(Z = 1|\mathbf{X} = \mathbf{x})) := f(\mathbf{x})$ where $f(\mathbf{x})$ is a centered Gaussian process. The considered prior Gaussian process is, most of the time, one-dimensional. Extensions to the multi-dimensional case include the so-called multi-tasks or multi-outputs GP models, see (Bonilla, Chai and Williams, 2007; Alvarez, Rosasco and Lawrence, 2012). Finally, some recent works focus on non Gaussian processes such as Student-t processes which have gain attention over the past years (Shah, Wilson and Ghahramani, 2014a; Chen, Wang and Gorban, 2020).

The discrete nature of Z makes the exact inference of model parameters infeasible. To overcome this difficulty, several techniques have been proposed, including the Laplace approximation, or through the expectation-propagation algorithm (Nickisch and Rasmussen, 2008). Such approaches rely on the inversion of a $n \times n$ covariance matrix and thus scale in $O(n^3)$ which makes the inference computationally demanding for large data sets. Scalable GPs were proposed to overcome this vexing effect, using for instance variational inference as in (Hensman, Matthews and Ghahramani, 2015). We refer to (Liu et al., 2020) for a review on this topic. In the next Section, we define a mixture of multivariate Gaussian processes which can be used for classification or imputation tasks without resort to approximate inference techniques.

3. Mixture of Multivariate Gaussian processes. The mixture of multivariate Gaussian processes model is introduced in Paragraph 3.1 and some associated properties are derived in Paragraph 3.2.

3.1. *Model.* Let \mathcal{T} be a compact subset of \mathbb{R} , throughout this document, we denote by $\mathcal{GP}_1(0, K)$ a continuous univariate centered Gaussian process on \mathcal{T} with covariance function $K : \mathcal{T}^2 \to \mathbb{R}$. Recall that, by definition, $W \sim \mathcal{GP}_1(0, K)$ implies that, for all $(t_1, \ldots, t_q) \in \mathcal{T}^q$, the random vector $(W(t_1), \ldots, W(t_q))^{\top}$ follows a multivariate centered Gaussian distribution $\mathcal{N}_q(\mathbf{0}, \Sigma)$ such that $\Sigma_{i,j} = K(t_i, t_j)$, see for instance (Rasmussen and Williams, 2006).

For all p > 0, let us similarly denote by $IGP_p(0, K)$ a p-dimensional, independent, centered Gaussian process defined as

$$\mathbf{W} = (W_1, \dots, W_p)^{\top} \sim I \mathcal{GP}_p(0, K) \text{ if and only if } \begin{cases} W_b \sim \mathcal{GP}_1(0, K), \ \forall b \in \{1, \dots, p\}, \\ W_b \perp W_{b'}, \ \forall b \neq b' \in \{1, \dots, p\}^2, \end{cases}$$

where \perp stands for independence. The above defined multivariate Gaussian processes are the building blocks to define more general multivariate Gaussian processes denoted by $\mathcal{MGP}_p(\mathbf{m}, K, \mathbf{A})$ where $\mathbf{m} : \mathcal{T} \to \mathbb{R}^p$ is the mean function, $K : \mathcal{T}^2 \to \mathbb{R}$ is the covariance operator and \mathbf{A} a non-singular $p \times p$ matrix:

(1)
$$\mathbf{Y} \sim \mathcal{MGP}_{p}(\mathbf{m}, K, \mathbf{A})$$
 if and only if $\mathbf{Y} = \mathbf{AW} + \mathbf{m}$ with $\mathbf{W} \sim \mathcal{IGP}_{p}(0, K)$.

Let us remark that model (1) is not identifiable without additional constraints. Indeed, $\mathcal{MGP}_p(\mathbf{m}, K, \mathbf{A})$ and $\mathcal{MGP}_p(\mathbf{m}, \lambda K, \mathbf{A}/\sqrt{\lambda})$ yield the same process for all $\lambda > 0$. This issue is discussed in further details in Section 4, see also the next paragraph for some basic properties of multivariate Gaussian processes defined in (1).

Introduce Z a discrete random variable taking its values in $c \in \{1, ..., C\}$ with $\pi_c = \mathbb{P}(Z = c)$. The mixture of multivariate Gaussian processes (M2GP) is defined by:

(2) Conditionally to
$$Z = c$$
, $\mathbf{Y} \sim \mathcal{MGP}_{p}(\mathbf{m}_{c}, K_{c}, \mathbf{A}_{c})$

where $\mathbf{m}_c : \mathcal{T} \to \mathbb{R}^p$, $K_c : \mathcal{T}^2 \to \mathbb{R}$ and \mathbf{A}_c is a non-singular $p \times p$ matrix, for all $c \in \{1, \dots, C\}$. In the context of SITS classification, **Y** represents the (unobserved) multidimensional process and p denotes the number of spectral bands. The particular case $\mathbf{A}_c = \mathbf{I}_p$ yields a mixture of independent Gaussian processes (MIGP) whose applications to classification have been investigated in (Constantin, Fauvel and Girard, 2021). Let us also note that multivariate Gaussian processes have already been used in the machine learning community, without formal definition though, see for instance the so-called multi-task Gaussian process (Bonilla, Chai and Williams, 2007) or the multivariate Gaussian process regression (Chen, Wang and Gorban, 2020).

3.2. *First properties.* Let **C** and **D** be two matrices of size $m \times n$ and $p \times q$ respectively. Recall that the Kronecker product $\mathbf{C} \otimes \mathbf{D}$ is the $mp \times nq$ matrix such that

$$\mathbf{C} \otimes \mathbf{D} = \begin{pmatrix} c_{11}\mathbf{D} \dots c_{n1}\mathbf{D} \\ \vdots & \ddots & \vdots \\ c_{m1}\mathbf{D} \dots c_{mn}\mathbf{D} \end{pmatrix}$$

and $vec(\mathbf{C}) \in \mathbb{R}^{mn}$ is the vector obtained by stacking the *n* columns of **C**:

$$\operatorname{vec}(\mathbf{C}) = (c_{11}, \dots, c_{m1}, c_{12}, \dots, c_{m2}, \dots, c_{1n}, \dots, c_{mn})^{\top}$$

Keeping these definitions in mind, the matrix-variate normal distribution $\mathcal{MN}_{p,q}$ (Dawid, 1981; Srivastava, von Rosen and von Rosen, 2008) is defined for all $p \times q$ random matrix \mathbf{Y}^* as:

(3) $\mathbf{Y}^{\star} \sim \mathcal{MN}_{p,q}(\mathbf{M}, \boldsymbol{\Sigma}, \boldsymbol{\Lambda})$ if and only if $\operatorname{vec}(\mathbf{Y}^{\star}) \sim \mathcal{N}_{pq}(\operatorname{vec}(\mathbf{M}), \boldsymbol{\Sigma} \otimes \boldsymbol{\Lambda})$,

where **M** is a $p \times q$ matrix, Σ and Λ are symmetric positive definite matrices of size $q \times q$ and $p \times p$ respectively. We refer to (Dawid, 1981) for an early definition of the matrix-variate normal distribution (as well as some of its derivatives), to (Gupta and Nagar, 1999) for a general account on matrix-variate distributions and to (Allen and Tibshirani, 2010) for an application to missing data imputation. The associated density function is defined for all $p \times q$ matrix **y** by

(4)
$$p(\mathbf{y}) = (2\pi)^{-pq/2} \det(\mathbf{\Sigma})^{-p/2} \det(\mathbf{\Lambda})^{-q/2} \exp\left(-\frac{1}{2} \operatorname{tr}\left[\mathbf{\Lambda}^{-1} (\mathbf{y} - \mathbf{M})\mathbf{\Sigma}^{-1} (\mathbf{y} - \mathbf{M})^{\mathsf{T}}\right]\right),$$

where $tr(\cdot)$ denotes the trace operator. The next Proposition establishes that the finite sized marginals of the multivariate Gaussian process (1) can be interpreted as random matrices from a matrix-variate normal distribution.

Proposition 1 Let $\mathbf{Y} \sim \mathcal{MGP}_p(\mathbf{m}, K, \mathbf{A})$ and introduce \mathbf{Y}^* the $p \times q$ random matrix defined as $\mathbf{Y}^* = (\mathbf{Y}(t_1), \dots, \mathbf{Y}(t_q))$ where $(t_1, \dots, t_q) \in \mathcal{T}^q$. Then,

(5)
$$\mathbf{Y}^{\star} \sim \mathcal{MN}_{p,q}(\mathbf{M}, \boldsymbol{\Sigma}, \mathbf{A}\mathbf{A}^{\top}),$$

where $\mathbf{M} = (\mathbf{m}(t_1), \dots, \mathbf{m}(t_q))$ and Σ is the covariance matrix defined by $\Sigma_{k,\ell} = K(t_k, t_\ell)$ for all $(k, \ell) \in \{1, \dots, q\}^2$. Equivalently,

$$vec(\mathbf{Y}^{\star}) \sim \mathcal{N}_{pq}(\boldsymbol{\mu}, \boldsymbol{\Sigma} \otimes \mathbf{A}\mathbf{A}^{\top}),$$

with $\mu = vec(\mathbf{M})$.

In the SITS framework, \mathbf{Y}^* represents the observed q-dimensional SITS which is a discretized version of \mathbf{Y} at q timestamps. An illustration is provided in Fig. 3 where $\mathcal{T} = [0, 1]$ and p = q = 10. Only the first two coordinates are represented for lack of space reasons. Let $\langle \cdot, \cdot \rangle$ denote the Euclidean scalar product on \mathbb{R}^p and $\|\cdot\|$ be the associated norm. For all non zero vectors $(u, v) \in \mathbb{R}^p \times \mathbb{R}^p$, we also introduce $\cos(u, v) = \langle u, v \rangle / (\|u\| \|v\|)$. As a direct consequence of the covariance structure in (5), the correlation ρ between the elements of the random matrix \mathbf{Y}^* can be derived:

Corollary 1 Suppose the assumptions of Proposition 1 hold.

(i) For all $(b, b') \in \{1, ..., p\}^2$ and $j \in \{1, ..., q\}$, one has

$$\rho(Y_{b,i}^{\star}, Y_{b',i}^{\star}) = \cos\left(\mathbf{a}_{b}, \mathbf{a}_{b'}\right)$$

(with \mathbf{a}_b the bth line of \mathbf{A}) and is thus independent of $j \in \{1, ..., q\}$. (ii) For all $(j, j') \in \{1, ..., q\}^2$ and $b \in \{1, ..., p\}$, one has

(6)
$$\rho(Y_{b,j}^{\star}, Y_{b,j'}^{\star}) = \Sigma_{j,j'} \left| \sqrt{\Sigma_{j,j} \Sigma_{j',j'}} \right|,$$

and is thus independent of $b \in \{1, ..., p\}$.

It appears that **A** tunes the dependence between the lines of \mathbf{Y}^{\star} (*i.e.* the spectral bands in the SITS context) while Σ drives the dependence between the columns (*i.e.* the acquisition times of the SITS).

A likelihood ratio test is introduced in (Lu and Zimmerman, 2005) to check whether the separability of the covariance (5) is adapted to the data in hand. However, this test has not been extended to irregularly sampled time-series. Let us also mention that, in (Mahanta, Aghaei and Plataniotis, 2013), the same Kronecker product model is used to regularize the estimation of the covariance matrix in high dimension.

4. Inference. This section addresses several inference aspects associated with the M2GP model. Consider $\{(\mathbf{Y}^1, Z_1), \dots, (\mathbf{Y}^n, Z_n)\}$ a set of *n* random pairs identically distributed from the M2GP model. Clearly, π_c can be estimated by its empirical counterpart $\hat{\pi}_c = n_c/n$ where $n_c = \sum_{i=1}^n \mathbb{I}\{Z_i = c\}$ is the number of samples in class *c* (and $\mathbb{I}\{\cdot\}$ is the indicator function). Besides, from (2), $\mathbf{Y}^i \sim \mathcal{MGP}_p(\mathbf{m}_c, K_c, \mathbf{A}_c)$ conditionally to $Z_i = c$, for all $i \in \{1, \dots, n\}$. The unknown quantities to be estimated are $\mathbf{m}_c : \mathcal{T} \to \mathbb{R}^p$, $K_c : \mathcal{T}^2 \to \mathbb{R}$ and the matrix \mathbf{A}_c . The use of parametric models for mean and covariance functions is discussed in Subsection 4.1 and the Maximum likelihood estimation (MLE) of all resulting parameters is presented in Subsection 4.2. The associated classification method based on the Maximum a posteriori (MAP) rule and the imputation of missing values are described in Subsection 4.3 and Subsection 4.4 respectively.

4.1. Parametric mean and covariance functions. Let J > 0 and introduce $\{\varphi_1, \ldots, \varphi_J\}$ a subset of J basis functions of $L_2(\mathcal{T})$. For all $b \in \{1, \ldots, p\}$, the *b*th coordinate $(\mathbf{m}_c(t))_b$ of $\mathbf{m}_c(t)$ is expanded as

(7)
$$(\mathbf{m}_{c}(t))_{b} = \sum_{j=1}^{J} \alpha_{c,b,j} \varphi_{j}(t),$$

with $t \in \mathcal{T}$, and where $a_{c,b,j}$ is the projection coefficient of $(\mathbf{m}_c(\cdot))_b$ on $\varphi_j(\cdot)$. Denoting by α_c the $p \times J$ matrix defined by:

$$\boldsymbol{\alpha}_{c} = \begin{pmatrix} \alpha_{c,1,1} & \alpha_{c,1,2} \dots & \alpha_{c,1,J} \\ \alpha_{c,2,1} & \ddots & \dots & \alpha_{c,2,J} \\ \vdots & \vdots & \ddots & \dots \\ \alpha_{c,p,1} & \dots & \dots & \alpha_{c,p,J} \end{pmatrix}$$

and letting $\mathbf{b} : t \in \mathcal{T} \mapsto (\varphi_1(t), \dots, \varphi_J(t))^\top \in \mathbb{R}^J$, then (7) can rewritten matricially as $\mathbf{m}_c(t) = \alpha_c \mathbf{b}(t)$.

The covariance operator K_c is assumed to belong to a family of symmetric positive-definite kernels, (Rasmussen and Williams, 2006, Chapter 4). A typical kernel is the squared exponential kernel (also known as Gaussian or RBF kernel) with an additive white noise covariance function:

(8)
$$K_c(t,t'|\boldsymbol{\theta}_c) = \gamma_c^2 \exp\left(-\frac{(t-t')^2}{2h_c^2}\right) + \sigma_c^2 \mathbb{I}\{t=t'\},$$

where $(t, t') \in \mathcal{T}^2$. The parameters are collected in θ_c with, in this case, $\theta_c = \{\gamma_c, h_c, \sigma_c\}$.

4.2. Maximum likelihood estimation. Assume each multivariate Gaussian process \mathbf{Y}^i is observed on its own finite grid of distinct q_i timestamps $(t_1^i, \ldots, t_{q_i}^i) \in \mathbb{R}^{q_i}$ and note $\mathbf{Y}^{i,\star} = (\mathbf{Y}^i(t_1^i), \ldots, \mathbf{Y}^i(t_{q_i}^i))^{\top}$ the associated $p \times q_i$ random matrix. Let us stress that this formalism naturally allows to deal with irregularly sampled SITS since the size of $\mathbf{Y}^{i,\star}$ may depend on *i*. From Proposition 1, one has

(9) Conditionally to
$$Z_i = c$$
, $\mathbf{Y}^{i,\star} \sim \mathcal{MN}_{p,q_i}(\boldsymbol{\alpha}_c \mathbf{B}^i, \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c), \mathbf{A}_c \mathbf{A}_c^{\top})$,

where the covariance matrix $\Sigma^{c,i}(\theta_c)$ is defined for all $(j, j') \in \{1, \dots, q_i\}^2$ by $\Sigma^{c,i}(\theta_c)_{j,j'} = K_c(t_j^i, t_j^i, |\theta_c)$ and $\mathbf{B}^i = (\mathbf{b}(t_1^i), \dots, \mathbf{b}(t_{q_i}^i))$ is a $J \times q_i$ design matrix. Parameters $\{\alpha_c, \theta_c, \mathbf{A}_c\}$ are estimated by minimizing the negative log-likelihood given hereafter.

Lemma 1 The negative log-likelihood associated with (9) can be expanded as

$$\mathcal{L} = \frac{1}{2} \sum_{c=1}^{C} \ell_c(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c, \mathbf{A}_c \mathbf{A}_c^{\top}),$$

(up to an additive constant) where, for all $c \in \{1, ..., C\}$,

(10)
$$\ell_{c}(\boldsymbol{\alpha}_{c},\boldsymbol{\theta}_{c},\mathbf{A}_{c}\mathbf{A}_{c}^{\top}) = Q_{c}\log\det(\mathbf{A}_{c}\mathbf{A}_{c}^{\top}) + p\sum_{i|Z_{i}=c}\log\det\Sigma^{c,i}(\boldsymbol{\theta}_{c}) + \operatorname{tr}\left(\sum_{i|Z_{i}=c}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})\{\Sigma^{c,i}(\boldsymbol{\theta}_{c})\}^{-1}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})^{\top}\{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1}\right),$$

with $Q_c = \sum_{i|Z_i=c} q_i$.

It appears that the likelihood only involves the product of matrices $\mathbf{A}_c \mathbf{A}_c^{\top}$ and not the matrix \mathbf{A}_c itself. This is a direct consequence of (5): The matrix-variate normal distribution of the sampled process \mathbf{Y}^* only depends on the above product. The parameters of interest are thus α_c , θ_c and $\mathbf{A}_c \mathbf{A}_c^{\top}$ and the MLE is obtained by solving *C* independent optimization problems:

(11)
$$(\hat{\alpha}_{c}, \hat{\theta}_{c}, \widehat{\mathbf{A}}_{c}^{\mathsf{T}}) = \underset{\boldsymbol{\alpha}_{c}, \boldsymbol{\theta}_{c}, \mathbf{A}_{c} \mathbf{A}_{c}^{\mathsf{T}}}{\arg\min \boldsymbol{\theta}_{c}, \boldsymbol{\theta}_{c}, \mathbf{A}_{c} \mathbf{A}_{c}^{\mathsf{T}})},$$

for all $c \in \{1, ..., C\}$. The solution is partially explicit as explained in the next Proposition.

Proposition 2 Let $c \in \{1, \ldots, C\}$.

(i) Solutions of (11) satisfy the following two properties. Given $\hat{\theta}_c$, one has:

(12)
$$\hat{\boldsymbol{\alpha}}_{c} = \left[\sum_{i|Z_{i}=c} \mathbf{Y}^{i,\star} \{\boldsymbol{\Sigma}^{c,i}(\hat{\boldsymbol{\theta}}_{c})\}^{-1} (\mathbf{B}^{i})^{\top}\right] \left[\sum_{i|Z_{i}=c} \mathbf{B}^{i} \{\boldsymbol{\Sigma}^{c,i}(\hat{\boldsymbol{\theta}}_{c})\}^{-1} (\mathbf{B}^{i})^{\top}\right]^{-1},$$

(13)
$$\widehat{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}} = \frac{1}{Q_{c}} \sum_{i|Z_{i}=c} (\mathbf{Y}^{i,\star} - \hat{\boldsymbol{\alpha}}_{c}\mathbf{B}^{i}) \{\boldsymbol{\Sigma}^{c,i}(\hat{\boldsymbol{\theta}}_{c})\}^{-1} (\mathbf{Y}^{i,\star} - \hat{\boldsymbol{\alpha}}_{c}\mathbf{B}^{i})^{\top}\}$$

(ii) The partial derivative of (10), w.r.t. the kth coordinate of θ_c is given by: (14)

$$\frac{\partial \ell_c(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c, \mathbf{A}_c \mathbf{A}_c^{\top})}{\partial (\boldsymbol{\theta}_c)_k} = \sum_{i \mid Z_i = c} \operatorname{tr} \left(\left[p\{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)\}^{-1} - \boldsymbol{\beta}^{c,i}(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c)^{\top} \{\mathbf{A}_c \mathbf{A}_c^{\top}\}^{-1} \boldsymbol{\beta}^{c,i}(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c) \right] \frac{\partial \boldsymbol{\Sigma}^{c,i}}{\partial (\boldsymbol{\theta}_c)_k} (\boldsymbol{\theta}_c) \right),$$

where $\boldsymbol{\beta}^{c,i}(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c) = (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)\}^{-1}.$

In practice, the computation of the MLE is achieved thanks to an iterative procedure based on (12)–(14), described in Algorithm 1 and discussed in Paragraph 4.5.

4.3. Supervised classification. Starting from a training set from the M2GP model $\{(\mathbf{Y}^{1,\star}, Z_1), \dots, (\mathbf{Y}^{n,\star}, Z_n)\}$ with $\mathbf{Y}^{i,\star} = (\mathbf{Y}^i(t_1^i), \dots, \mathbf{Y}^i(t_{q_i}^i))^\top$, our goal is to assign a label $\tilde{c} \in \{1, \dots, C\}$ to a new $p \times q$ random matrix $\tilde{\mathbf{Y}}^{\star} = (\mathbf{Y}(\tilde{t}_1), \dots, \mathbf{Y}(\tilde{t}_q))^\top$. We focus on the MAP rule which consists in maximizing w.r.t. c the posterior probability

$$\mathbb{P}(Z=c|\tilde{\mathbf{Y}}^{\star}) \propto \pi_c p(\tilde{\mathbf{Y}}^{\star}|Z=c),$$

where $p(\tilde{\mathbf{Y}}^{\star}|Z=c)$ is matrix-variate normal density defined as

(15)
$$p(\tilde{\mathbf{Y}}^{\star}|Z=c) = (2\pi)^{-pq/2} \det\left(\tilde{\mathbf{\Sigma}}^{c}(\boldsymbol{\theta}_{c})\right)^{-p/2} \det\left(\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\right)^{-q/2} \\ \times \exp\left(-\frac{1}{2}\mathrm{tr}\left[\{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1}(\tilde{\mathbf{Y}}^{\star}-\boldsymbol{\alpha}_{c}\tilde{\mathbf{B}})\tilde{\mathbf{\Sigma}}^{c}(\boldsymbol{\theta}_{c})^{-1}(\tilde{\mathbf{Y}}^{\star}-\boldsymbol{\alpha}_{c}\tilde{\mathbf{B}})^{\top}\right]\right),$$

see (22) in the proof of Lemma 1. Here, the covariance matrix $\tilde{\Sigma}^{c}(\theta_{c})$ is defined for all $(j, j') \in \{1, ..., q\}^{2}$ by $\tilde{\Sigma}^{c}(\theta_{c})_{j,j'} = K_{c}(\tilde{t}_{j}, \tilde{t}_{j'}|\theta_{c})$ and $\tilde{\mathbf{B}} = (\mathbf{b}(\tilde{t}_{1}), ..., \mathbf{b}(\tilde{t}_{q}))$ is a $J \times q$ design matrix. In practice, all parameters are replaced using their MLE counterparts and \tilde{c} is selected by minimizing the negative log posterior probability, that is:

$$\tilde{c} = \arg\min_{c} \left\{ p \log \det\left(\tilde{\boldsymbol{\Sigma}}^{c}(\hat{\boldsymbol{\theta}}_{c})\right) + q \log \det\left(\widehat{\boldsymbol{A}_{c}}\mathbf{A}_{c}^{\top}\right) - 2 \log(n_{c}/n) + \operatorname{tr}\left[\left\{\widehat{\boldsymbol{A}_{c}}\mathbf{A}_{c}^{\top}\right\}^{-1}(\tilde{\mathbf{Y}}^{\star} - \hat{\boldsymbol{\alpha}}_{c}\tilde{\mathbf{B}})\tilde{\boldsymbol{\Sigma}}^{c}(\hat{\boldsymbol{\theta}}_{c})^{-1}(\tilde{\mathbf{Y}}^{\star} - \hat{\boldsymbol{\alpha}}_{c}\tilde{\mathbf{B}})^{\top}\right] \right\}.$$

In the SITS framework, the above formula provides a natural way to classify a new multivariate time-series even though it is not observed at the same timestamps as the examples from the training set.

4.4. Imputation of missing values. The next result provides the distribution of the MGP process at time t^{\dagger} conditionally to its label and to observations at times t_1, \ldots, t_q .

Proposition 3 Assume that, conditionally to Z = c, $\mathbf{Y} \sim \mathcal{MGP}_p(\boldsymbol{\alpha}_c \mathbf{b}, K_c, \mathbf{A}_c)$ and introduce \mathbf{Y}^* the $p \times q$ random matrix defined as $\mathbf{Y}^* = (\mathbf{Y}(t_1), \dots, \mathbf{Y}(t_q))$ where $(t_1, \dots, t_q) \in \mathcal{T}^q$. Let $t^{\dagger} \in \mathcal{T}$ such that $t^{\dagger} \neq t_k$ for all $k \in \{1, \dots, q\}$. Then,

conditionally to
$$Z = c$$
 and $\mathbf{Y}^{\star} = \mathbf{y}^{\star}, \ \mathbf{Y}(t^{\dagger}) \sim \mathcal{N}_p(\boldsymbol{\mu}_c(t^{\dagger}, \mathbf{y}^{\star}), \ \boldsymbol{\Lambda}_c(t^{\dagger}))$

with

$$\boldsymbol{\mu}_{c}(t^{\dagger}, \mathbf{y}^{\star}) = \boldsymbol{\alpha}_{c} \mathbf{b}(t^{\dagger}) + (\mathbf{y}^{\star} - \boldsymbol{\alpha}_{c} \mathbf{B}) \{\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})\}^{-1} \mathbf{k}_{c}(t^{\dagger}),$$
$$\boldsymbol{\Lambda}_{c}(t^{\dagger}) = \left[K_{c}(t^{\dagger}, t^{\dagger} | \boldsymbol{\theta}_{c}) - \mathbf{k}_{c}(t^{\dagger})^{\top} \boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1} \mathbf{k}_{c}(t^{\dagger})\right] \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top}$$

and where $\mathbf{k}_c(t^{\dagger}) = (K_c(t^{\dagger}, t_1 | \boldsymbol{\theta}_c), \dots, K_c(t^{\dagger}, t_q | \boldsymbol{\theta}_c))^{\top}$. Recall that the covariance matrix $\boldsymbol{\Sigma}^c(\boldsymbol{\theta}_c)$ is defined for all $(j, j') \in \{1, \dots, q\}^2$ by $\boldsymbol{\Sigma}^c(\boldsymbol{\theta}_c)_{j,j'} = K_c(t_j, t_{j'} | \boldsymbol{\theta}_c)$ and $\mathbf{B} = (\mathbf{b}(t_1), \dots, \mathbf{b}(t_q))$ is a $J \times q$ design matrix.

As a consequence, when $\mathbf{Y}(t^{\dagger})$ is not observed (but its label is known to be *c*), this missing value can be imputed by the conditional expectation given in Proposition 3, where the unknown parameters are replaced by their associated MLE:

(16)
$$\hat{\mathbf{Y}}_{c}(t^{\dagger}) = \hat{\alpha}_{c}\mathbf{b}(t^{\dagger}) + (\mathbf{Y}^{\star} - \hat{\alpha}_{c}\mathbf{B})\{\boldsymbol{\Sigma}^{c}(\hat{\boldsymbol{\theta}}_{c})\}^{-1}\hat{\mathbf{k}}_{c}(t^{\dagger}).$$

This allows for the reconstruction of SITS values at unobserved times. If the label of \mathbf{Y}^{\star} is unknown, the distribution of the MGP process at time t^{\dagger} conditionally to observations at times t_1, \ldots, t_q can still be derived from Proposition 3:

conditionally to
$$\mathbf{Y}^{\star} = \mathbf{y}^{\star}, \ \mathbf{Y}(t^{\dagger}) \sim \sum_{c=1}^{C} \mathbb{P}(Z = c | \mathbf{Y}^{\star} = \mathbf{y}^{\star}) \mathcal{N}_{p} \left(\boldsymbol{\mu}_{c}(t^{\dagger}, \mathbf{y}^{\star}), \ \boldsymbol{\Lambda}_{c}(t^{\dagger}) \right),$$

leading to

$$\boldsymbol{\mu}(t^{\dagger}, \mathbf{y}^{\star}) = \sum_{c=1}^{C} \mathbb{P}(Z = c | \mathbf{Y}^{\star} = \mathbf{y}^{\star}) \boldsymbol{\mu}_{c}(t^{\dagger}, \mathbf{y}^{\star}),$$
$$\boldsymbol{\Lambda}(t^{\dagger}) = \sum_{c=1}^{C} \mathbb{P}(Z = c | \mathbf{Y}^{\star} = \mathbf{y}^{\star}) \left(\boldsymbol{\Lambda}_{c}(t^{\dagger}) + \boldsymbol{\mu}_{c}(t^{\dagger}, \mathbf{y}^{\star})^{\mathsf{T}} \boldsymbol{\mu}_{c}(t^{\dagger}, \mathbf{y}^{\star}) \right) - \boldsymbol{\mu}(t^{\dagger}, \mathbf{y}^{\star})^{\mathsf{T}} \boldsymbol{\mu}(t^{\dagger}, \mathbf{y}^{\star}).$$

Thus, when both $\mathbf{Y}(t^{\dagger})$ and its label are not observed, $\mathbf{Y}(t^{\dagger})$ can be imputed by

(17)
$$\hat{\mathbf{Y}}(t^{\dagger}) = \sum_{c=1}^{C} \widehat{\mathbb{P}}(Z = c | \mathbf{Y}^{\star}) \hat{\mathbf{Y}}_{c}(t^{\dagger}),$$

where $\hat{\mathbf{Y}}_{c}(t^{\dagger})$ is given in (16), $\mathbf{Y}^{\star} = (\mathbf{Y}(t_{1}), \dots, \mathbf{Y}(t_{q}))$ and

$$\widehat{\mathbb{P}}(Z=c|\mathbf{Y}^{\star}) = \hat{\pi}_{c}\hat{p}(\mathbf{Y}^{\star}|Z=c) \left| \sum_{k=1}^{C} \hat{\pi}_{k}\hat{p}(\mathbf{Y}^{\star}|Z=k) \right|,$$

with $\hat{p}(\mathbf{Y}^{\star}|Z=k)$ the estimated matrix-variate density defined similarly to (15) by

$$\hat{p}(\mathbf{Y}^{\star}|Z=k) = (2\pi)^{-pq/2} \det\left(\boldsymbol{\Sigma}^{k}(\hat{\boldsymbol{\theta}}_{k})\right)^{-p/2} \det\left(\widehat{\mathbf{A}_{k}\mathbf{A}_{k}^{\top}}\right)^{-q/2} \\ \times \exp\left(-\frac{1}{2} \operatorname{tr}\left[\widehat{\mathbf{A}_{k}\mathbf{A}_{k}^{\top}}\right]^{-1} (\mathbf{Y}^{\star} - \hat{\boldsymbol{\alpha}}_{k}\mathbf{B})\boldsymbol{\Sigma}^{c}(\hat{\boldsymbol{\theta}}_{k})^{-1} (\mathbf{Y}^{\star} - \hat{\boldsymbol{\alpha}}_{k}\mathbf{B})^{\top}\right]\right).$$

4.5. Numerical implementation. The computation of the MLE is implemented as detailed in Algorithm 1 using the results of Proposition 2. To deal with the idenfiability issue mentioned in Paragraph 3.1, $\mathbf{A}_c \mathbf{A}_c^{\top}$ is normalised by η_c such that $\|\mathbf{A}_c \mathbf{A}_c^{\top}\|_F = 1$ (where $\|\cdot\|_F$ denotes the Frobenius norm) and each covariance matrix $\Sigma^{c,i}(\theta_c)$ is modified accordingly so that the likelihood remains unaffected (step (d) of Algorithm 1). The gradient step (e) is performed using the Broyden-Fletcher-Goldfarb-Shanno (BFGS) algorithm, see (Zhu et al., 1997). More specifically, the L-BFGS-B version is used which allows for box and positivity constraints. As described in (Zhu et al., 1997), the gradient step is obtained by line search and the algorithm stops when: the objective function (*i.e.* the likelihood) does not change significantly, the (infinite) norm of the projected gradient is sufficiently small or when the maximum number of iterations is reached. Since the objective function is not convex, the optimization process is sensitive to the initialization. In practice, multiple random restarts are used and the best solution is retained. Let us highlight that, in practice, steps (a)-(e) are computed for all classes in parallel since the model parameters are decoupled w.r.t. the classes.

Input : Sample $\{(\mathbf{Y}^{i,\star}, Z_i) \in \mathbb{R}^{p \times q_i} \times \{1, \dots, C\}, i = 1, \dots, n\}$ and initialization $(\theta_1, \dots, \theta_C)$. Output: MLE $(\hat{\alpha}_c, \widehat{\mathbf{A}_c \mathbf{A}_c^{\top}}, \hat{\theta}_c), c = 1, \dots, C$. for c = l to C do repeat (a) Update α_c using (12); (b) Update $\mathbf{A}_c \mathbf{A}_c^{\top}$ using (13); (c) Compute $\eta_c \leftarrow ||\mathbf{A}_c \mathbf{A}_c^{\top}||_F$; (d) Update $\mathbf{A}_c \mathbf{A}_c^{\top} \leftarrow \mathbf{A}_c \mathbf{A}_c^{\top} / \eta_c$ and $\Sigma^{c,i}(\theta_c) \leftarrow \eta_c \Sigma^{c,i}(\theta_c), i = 1, \dots, n$; (e) Update θ_c with a gradient step using (14); until $\ell_c(\alpha_c, \mathbf{A}_c \mathbf{A}_c^{\top}, \theta_c)$ has converged; end

Algorithm 1: Computation of MLE of model parameters.

The numerical complexity of one iteration for all classes of Algorithm 1 is $O(n(q_{\infty}^3 + p^3 + J^3))$ where *n* is the sample size and $q_{\infty} = \max\{q_i, i = 1, ..., n\}$. The computation of the MLE thus scales linearly w.r.t. *n*. In constrast, the cost associated with standard classification methods based on Gaussian processes is $O((C + 1)n^3)$ (Rasmussen and Williams, 2006, Algorithm 3.3). Here, the computation of the MLE only relies on the inversion of $p \times p$ and $q_i \times q_i$ matrices whose sizes do not depend on the sample size.

Let us note that Algorithm 1 can be interpreted as an extension of the so-called Flip-flop method introduced independently by Mardia and Goodall (1993); Dutilleul (1999). This latter method in an iterative way to compute the MLE associated with the matrix-variate normal distribution. As such, it is limited to the situation where $q_1 = q_2 = \cdots = q_n$ which only occurs

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when all Gaussian processes are observed on a common grid. Identifiability issues are discussed in (Srivastava, von Rosen and von Rosen, 2008) and the method is extended to higher order tensor distributions in (Manceur and Dutilleul, 2013). Applications of matrix-variate normal distribution are found in different contexts such as electro-encephalography (Spinnato et al., 2014) or remote sensing (Glanz and Carvalho, 2018).

Finally, all the above estimations procedures have been implemented in Python using the Scikit-Lean API, see (Buitinck et al., 2013). The Fourier basis $\{\varphi_1, \ldots, \varphi_J\}$ was chosen to estimate the mean function (see (Constantin, Fauvel and Girard, 2021) for other bases), while the family of symmetric positive-definite kernels was selected among the *Kernels* class in the Scikit-Learn library.

5. Validation on simulated data. The performance of the inference procedure associated with the M2GP model is illustrated on simulated data.¹ The simulated model is described in Paragraph 5.1. First, the influence of the dependence between coordinates as well as the influence of the number of observation times are investigated in Paragraph 5.2. Second; the consequences on the classification and imputation accuracy are discussed in Paragraph 5.3.

5.1. Experimental design. A binary classification problem is considered. Two classes are simulated from a 10-dimensional M2GP model on $\mathcal{T} = [0, 1]$ with 1,000 samples per class leading to n = 2,000 and p = 10. Mean functions are generated following (7) with a Fourier basis of size J = 11. Coefficients $\alpha_{c,b,j}$ are simulated independently from a $\mathcal{N}_1(0,0.02)$ distribution, $c \in \{1,2\}$, $b \in \{1,\ldots,10\}$ and $j \in \{1,\ldots,11\}$. The covariance operator is identical for both classes: $K_1(\cdot, \cdot) = K_2(\cdot, \cdot)$. It is defined following (8) as the sum of a RBF kernel and a white noise covariance function. The associated parameters are $\theta_1 = \{\gamma_1, h_1, \sigma_1\} = \{1.5, 150, 0.05\} = \theta_2$. We also set $\mathbf{A}_1 = \mathbf{A}_2$ with

(18)
$$\mathbf{A}_{1}\mathbf{A}_{1}^{\mathsf{T}} = \begin{pmatrix} 1 \ \beta \ \cdots \ \beta \\ \beta \ 1 \ \cdots \ \vdots \\ \vdots \ \vdots \ \ddots \ \beta \\ \beta \ \cdots \ \beta \ 1 \end{pmatrix},$$

so that β tunes the pairwise correlation between the 10 coordinates of the Gaussian processes. In the following, we shall consider $\beta \in \{0, 1/4, 1/2\}$. In practice, M2GP processes are simulated on random grids of varying size $q \in \{10, 20, ..., 100\}$, see Fig. 3 for an illustration in the case q = 10 and $\beta = 0$.

5.2. *Estimation results*. All estimation procedures are evaluated on 100 replications of the above described simulation model. First, for all $c \in \{1, 2\}$, the quality of the reconstructed mean $\hat{\mathbf{M}}_c = \hat{\alpha}_c \mathbf{B}$ is measured by the normalized Mean Squared Error (nMSE) defined as:

(19)
$$\mathrm{nMSE}(\hat{\mathbf{M}}_{c},\mathbf{M}_{c}) = \frac{\|\mathbf{M}_{c} - \hat{\mathbf{M}}_{c}\|_{F}^{2}}{\|\mathbf{M}_{c} - \overline{\mathbf{M}}_{c}\|_{F}^{2}},$$

where $\overline{\mathbf{M}}_c$ is the empirical mean of the processes in class *c*. The lower this score is, the better the estimation. An example of reconstructed mean is presented on Fig. 4, for one replication.

¹The code and a notebook are available at https://gitlab.inria.fr/aconstan/mixture-ofmultivariate-gaussian-processes-for-classification-of-irregularly-sampled-satelliteimage-time-series.



Fig 3. Two simulated M2GP processes (transparent lines) in dimension p = 10 observed at q = 10 timestamps (dots), from two classes (c = 1: blue, c = 2: red). The mean functions are depicted as continuous opaque lines. Top panel: first coordinates, bottom panel: second coordinates (only the first two coordinates p_1 and p_2 are represented).

Second, the quality of the estimation of the covariance structure $\mathbf{A}_c \mathbf{A}_c^{\mathsf{T}}$ (see 18) by $\widehat{\mathbf{A}_c \mathbf{A}_c^{\mathsf{T}}}$ is assessed by the cosine score defined as:

(20)
$$C(\widehat{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}},\mathbf{A}_{c}\mathbf{A}_{c}^{\top}) = 1 - \frac{\langle \widehat{\mathbf{A}_{c}}\mathbf{A}_{c}^{\top},\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\rangle_{F}}{\|\widehat{\mathbf{A}_{c}}\mathbf{A}_{c}^{\top}\|_{F}\|\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\|_{F}}$$

Let us note that $C(\widehat{\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}}, \mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}) \in [0, 2]$ with $C(\widehat{\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}}, \mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}) = 0$ when $\widehat{\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}}$ and $\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}$ are proportional. Finally, turning to the estimation of the kernel part (8) of the dependence structure, we focus on the estimation accuracy of the length-scale by computing the absolute difference between the true length-scale $h_1 = h_2 = 150$ and its estimated counterpart. The results are averaged over the 100 independent replications and are reported on Fig. 5 for the first class. Similar results are obtained for the second one. It appears that, unsurprisingly, the quality of the estimates increases with the number q of discretization times. At the opposite, the dependence parameter β does not seem to influence much the accuracy of the estimation. One can nevertheless note that, as expected, the variability of the estimated sincreases with β , as the information carried by correlated coordinates decreases. Besides, the estimated length-scales do not depend on β , this may be explained by the separability property exhibited in Corollary 1.

5.3. Classification and imputation results. Here, we focus on the comparison between results associated with M2GP and MIGP models. To assess the classification and imputation performances, 4,000 samples are generated following the model described in Paragraph 5.1 and then split into two disjoint balanced sets. The first one is used as a training set (of size n = 2,000) to estimate model parameters. The second one is used as a test set where the



Fig 4. Estimation of mean functions by M2GP on simulated data for all coordinates $b \in \{1, ..., 10\}$, classes $c \in \{1, 2\}$ and $\beta = 0$ on one replication. The dashed line is the true mean, the red line is the estimated GP mean from a discretization on a grid of size q = 10.





Fig 5. Estimation of M2GP parameters on simulated data as a function of the number q of discretization times on class c = 1. From left to right: normalized mean squared error (19), cosine score (20) and absolute difference of length-scales. From top to bottom: $\beta = 0$, $\beta = 1/4$ and $\beta = 1/2$.

accuracy of the classification and imputation steps associated with the two above methods are compared. The classification performance is assessed thanks to the Overall Accuracy (OA), that is the ratio of the number correctly classified test observations and the total number of test observations, while the nMSE is used for the imputation task. Similarly to (19), we let

(21)
$$\operatorname{nMSE}(\hat{\mathbf{Y}}^{\star}, \mathbf{Y}^{\star}) = \frac{\|\hat{\mathbf{Y}}^{\star} - \mathbf{Y}^{\star}\|_{F}^{2}}{\|\mathbf{Y}^{\star} - \overline{\mathbf{Y}^{\star}}\|_{F}^{2}}$$

where $\hat{\mathbf{Y}}^{\star}$ is the imputed discretized process when the class is unknown thanks to (17), given the observed discretized process on q points. $\overline{\mathbf{Y}^{\star}}$ is the empirical mean of discretized processes in the test set. The above Frobenius norms are computed on a fixed regular grid of \mathcal{T} defined as { $t_{\ell} = \ell/100, \ell = 1, ..., 100$ }. The results are reported in Fig. 6.

It appears that the classification scores associated with M2GP increase with the dependence coefficient β and the number q of discretization times. On the opposite, MIGP scores are decreasing with β , due to the independence assumption. When there is no dependence



Fig 6. Classification overall accuracy (OA, left panel) and reconstruction normalized mean-squared error (nMSE, right panel in log scale) boxplots computed on simulated data. Comparison between M2GP (blue) and MIGP (red) results as functions of the number q of discretization times. From top to bottom: $\beta = 0$, $\beta = 1/4$ and $\beta = 1/2$.

between coordinates ($\beta = 0$), both methods provide similar classification scores. Unsurprisingly, M2GP outperforms MIGP as soon as a dependence occurs.

In terms of reconstruction, both methods feature similar performances, increasing with q. The dependence strength only impacts the variance of the reconstructed processes: The larger β is, the larger the variability.

6. Time-series classification: Application to satellite data. This section is devoted to multivariate SITS classification using the M2GP model. The data were acquired by the Sentinel-2 satellite, and are presented in Paragraph 6.1, with a focus on the irregular temporal sampling. The estimated M2GP parameters are interpreted and discussed in Paragraph 6.2. Finally Paragraph 6.3 concludes this section with classification results and comparisons to state-of-the-art methods.

6.1. Sentinel-2 satellite image time-series. Since 2016, the Sentinel-2 mission (Drusch et al., 2012) produces massive multispectral images,² around 1.6TBytes a day, with a spatial resolution of 10 m/pixel and 13 spectral bands (only 10 bands are used for the analysis). The frequency of revisit is 5 days and clouds as well as shadows are present in the data, at random locations. Most of the clouds and shadows positions are automatically extracted by the data provider. Yet, thin clouds may remain in the data. The selected images cover the area of Toulouse, France (Fig. 7) and all available acquisitions for the year 2018 were used. The image is of spatial size 10,000×10,000 pixels (10,000 km²). Each extracted time-series *i* has a dimension of p = 10 channels (or bands) and its own number of timestamps q_i . The distribution of the q_i s is represented in Fig. 8 for this area in 2018.

The supervised classification task consists in assigning a pre-defined label to every pixel of the image. Fourteen classes were extracted from national data-bases and 10 pairs of training and validation data-sets are generated independently for the experiments by randomly selecting samples for the training and testing sets. Training and testing sets were carefully constructed to avoid spatial dependence between pixels.

Table 1 shows the number of extracted samples for each training and validation set. The number of samples per class is unbalanced but represents the actual proportion of land cover classes in the region.

6.2. Parameters estimation. M2GP is fitted to the satellite image time-series using the estimators described in Section 4. A Fourier basis is adopted for estimating the means using J = 19 functions while the time dependence structure is modeled by a RBF kernel combined with an additive white noise. The choice of the basis and the selection of the dimension J are discussed in the MIGP framework by (Constantin, Fauvel and Girard, 2021, Fig. 8, and Fig. 1 in the supp. mat.).

Estimated mean functions are reported in Fig. 9 for four selected channels: blue, green, red and near infrared (nIR) and four selected classes: continuous urban fabric, summer crops, broad-leaved forest and water bodies. In the context of remote sensing data, nIR is often correlated with the presence or absence of vegetation: Large values of nIR associated with small values of red, indicate that the vegetation is abundant. This behavior is observed in agricultural classes such as summer crops or broad-leaved forest during spring and summer.

The estimated covariance matrices between all 10 channels $\mathbf{A}_c \mathbf{A}_c^{\mathsf{T}}$ are reported in Fig. 10 for the same classes. Similar covariance matrices have already been observed on mono-temporal Sentinel-2 data, we refer to (Wang et al., 2018, Fig. 8) for similar results on crops classes.

²https://sentinel.esa.int/web/sentinel/missions/sentinel-2/data-products.



FIG 7. The study area is located in the south of France (right bottom image). The left bottom image corresponds to the entire area ($100 \text{ km} \times 100 \text{ km}$) and the upper image is a zoom over the red rectangle ($11 \text{ km} \times 5 \text{ km}$).



FIG 8. Normalized histogram of the q_i s within the SITS data-set.

Finally, the time covariance structure is illustrated on Fig. 11. The estimated RBF kernel on the same four classes is drawn when centered at day 180. The temporal correlation associated with natural elements, such as summer crops or broad-leaved forest, is short since their reflectance evolves along the year (*e.g.* because of the vegetation cycle, or anthropic events). In contrast, man-made materials, such as continuous urban fabrics, exhibit longer temporal correlation because their reflectance does not evolve along the time.³

6.3. *Classification results*. In this section, the classification performances of M2GP are compared to state-of-the-art methods. Four competitors are considered: Random forests

³This is true when the period of observation is not too long, few years, otherwise the material property might be altered and its reflectance could vary.



FIG 9. Estimated means for four channels and four classes (continuous urban fabric, summer crops, broad-leaved forest and water bodies). The horizontal axis represents the days of the year and the vertical axis represents the reflectance value.



FIG 10. Estimated covariance matrices $\mathbf{A}_{c}\mathbf{A}_{c}^{\top}$ on four land-cover classes.

Table 1
Land cover classes and number of extracted samples n_c per class for each training and validation se

Class	n_c
Summer crops	40,000
Winter crops	30,000
Broad-leaved forest	10,000
Continuous urban fabric	10,000
Discontinuous urban fabric	10,000
Industrial or commercial units	10,000
Meadow	10,000
Orchards	10,000
Road surfaces	10,000
Vines	10,000
Water bodies	10,000
Woody moorlands	9,972
Coniferous forest	9,957
Natural grasslands	9,939
Total	189,868



Fig 11. Normalized RBF kernels (8) centered at day 180: $K(t, 180) = \exp(-0.5(t - 180)^2/h_c^2)$ computed on four classes.

(RF) (Breiman, 2001), Quadratic discriminant analysis (QDA) which is based on a finitedimensional Gaussian model, linear Support vector machine (SVM) classifier fitted with a Stochastic Gradient Descent (Zhang, 2004), and, finally, Mixture of independent Gaussian processes (MIGP) (Constantin, Fauvel and Girard, 2021)).

The time-series have been resampled on a common temporal grid of size 73 (every 5 days of year 2018) using a linear interpolation for RF, QDA and SVM methods since they require a fix vectorial representation of the sample. All the spectral bands have been stacked together to obtain a vector of dimension 73 dates × 10 spectral bands = 730 features. RF is trained with 100 trees of depth 25, and QDA is used with a regularized version of the estimated covariance matrix (Friedman, 1989), $\tilde{\Sigma} = (1 - \epsilon)\hat{\Sigma} + \epsilon \mathbf{I}$, with $\epsilon = 10^{-2}$.

The F_1 -score is computed to assess numerically the classification accuracy. The F_1 is defined as the harmonic mean of the precision and recall scores (Tharwat, 2021). Classification maps are also presented in order to qualitatively evaluate the spatial coherency of the results (despite a spatial pixel-wise independence assumption made by all considered methods).

	QDA	RF	SVM	MIGP	M2GP
Summer crops	96.5 ± 0.27	96.8 ± 0.45	95.6 ± 0.81	90.0 ± 0.83	95.9 ± 0.44
Winter crops	91.6 ± 0.48	94.0 ± 0.77	93.9 ± 0.66	80.2 ± 0.83	92.2 ± 0.64
Broad-leaved forest	77.4 ± 3.91	86.2 ± 2.35	85.3 ± 2.63	75.7 ± 5.03	81.5 ± 3.10
Cont. urban fabric	39.8 ± 6.18	58.0 ± 1.55	55.9 ± 2.49	21.4 ± 3.49	30.9 ± 5.51
Discont. urban fabric	58.5 ± 1.39	57.3 ± 3.44	40.2 ± 12.61	42.5 ± 3.17	54.5 ± 0.80
Ind. or commercial units	31.3 ± 2.14	60.3 ± 1.35	48.3 ± 4.05	27.4 ± 0.92	38.4 ± 2.34
Meadow	58.3 ± 4.14	64.8 ± 2.94	63.0 ± 3.17	43.3 ± 3.80	55.0 ± 4.19
Orchards	72.9 ± 4.05	81.0 ± 2.64	76.4 ± 3.11	51.9 ± 5.46	77.6 ± 3.58
Road surfaces	73.1 ± 1.92	87.1 ± 1.87	78.7 ± 2.79	54.2 ± 5.79	75.0 ± 2.06
Vines	71.1 ± 4.35	78.9 ± 6.86	78.5 ± 6.57	60.9 ± 7.61	71.7 ± 5.18
Water bodies	98.7 ± 0.35	99.4 ± 0.08	99.3 ± 0.10	84.9 ± 5.38	96.8 ± 0.84
Woody moorlands	23.9 ± 7.70	56.6 ± 3.50	56.1 ± 3.85	14.1 ± 5.52	10.6 ± 12.00
Coniferous forest	76.6 ± 7.24	86.9 ± 2.76	87.0 ± 2.56	61.2 ± 5.41	82.4 ± 6.61
Natural grasslands	29.8 ± 12.88	30.7 ± 16.90	19.4 ± 14.68	15.4 ± 7.86	20.6 ± 8.46
Average F_1 score	70.5 ± 0.75	78.2 ± 1.17	75.2 ± 1.11	57.4 ± 1.04	70.1 ± 0.43

TABLE 2 Mean F_1 score (mean(%) ± standard deviation) on the 10 independent data-sets.

Means F_1 scores and their standard deviations computed on 10 independent sets are reported in Table 2 for each class as well as the "average F_1 score" computed on all classes. Nnon-parametric methods (RF and SVM) provide the best classification results in terms of F1-score. The uni-modal assumption induced by Gaussian models may thus be ill-adapted to this data-set. M2GP and QDA provide lower and similar accuracy, even though M2GP is based on stronger assumptions on the covariance structure than QDA.

The obtained classification maps are reported in Fig. 12 for 3 different sites. Large differences are observed in these scenes. For the first column, corresponding to the airport zone, most of the inner vegetations are wrongly classified to natural grasslands with QDA, while RF, SVM and M2GP classify correctly them as meadow. Runway are mostly confused with industrial/commercial units using RF while runways are almost recovered by M2GP. Overall, strong differences between thematic maps are observed, but visual assessment from a monodate color image is difficult. Yet, without taking into account the spatial dependence, M2GP recovers most of the spatial structure of the image, and the *salt and pepper* classification noise is limited, as for RF and SVM.

7. Discussion. A multivariate Gaussian process model has been introduced for the classification of irregularly sampling satellite image time-series. The multivariate model involves a specific structure of the covariance operator that exploits the data features and also reduces the number of parameters to estimate. Furthermore, the proposed formulation scales linearly w.r.t. the number of samples. Experimental results on simulated and real data sets show the importance of modeling the dependence between coordinates of the process, in particular for classification accuracy.

Current development concerns the use of two satellite sources. Sentinel-2 satellites are complemented with Sentinel-1 ones (which are not affected by clouds) which acquire radar data (with a different physical content): An extension of the proposed model will consist in combining these two time-series with irregular temporal and spectral sampling. Another possible extension would be to consider a non spatially stationary mean function, as in (Cressie, 1993).

Finally, these models can be extended to non-Gaussian processes, *e.g.* Student-t as in (Shah, Wilson and Ghahramani, 2014b; Chen, Wang and Gorban, 2020) and applied to the unsupervised classification problem.



Fig 12. Three extracts of the classification maps obtained by QDA, RF, SVM, MIGP and M2PG methods.

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APPENDIX: PROOFS

Proof of Proposition 1. Let $\mathbf{Y} \sim \mathcal{MGP}_p(\mathbf{m}, K, \mathbf{A})$ and introduce \mathbf{Y}^* the $p \times q$ random matrix defined as $\mathbf{Y}^* = (\mathbf{Y}(t_1), \dots, \mathbf{Y}(t_q))$ where $(t_1, \dots, t_q) \in \mathcal{T}^q$. From (1), we have $\mathbf{Y} = \mathbf{AW} + \mathbf{m}$ with $\mathbf{W} \sim \mathcal{IGP}_p(\mathbf{0}, K)$. Let $\mathbf{W}^* = (\mathbf{W}(t_1), \dots, \mathbf{W}(t_q))$ be the associated $p \times q$ random matrix. Our first goal is to prove that $\mathbf{W}^* \sim \mathcal{MN}_{p,q}(\mathbf{0}, \boldsymbol{\Sigma}, \mathbf{I}_p)$ or, equivalently, from (3), to prove that $\operatorname{vec}(\mathbf{W}^*) \sim \mathcal{N}_{pq}(\mathbf{0}, \boldsymbol{\Sigma} \otimes \mathbf{I}_p)$. To this end, let us consider the random variable

$$S = \sum_{b=1}^{p} \sum_{j=1}^{q} \lambda_{b,j} \mathbf{W}_{b,j}^{\star}$$

and let us prove that S is a Gaussian random variable pour all $\lambda_{b,i} \in \mathbb{R}^{pq}$. Clear, one also has

$$S = \sum_{b=1}^{p} S_{b}, \text{ with } S_{b} := \sum_{j=1}^{q} \lambda_{b,j} \mathbf{W}_{b,j}^{\star} = \sum_{j=1}^{q} \lambda_{b,j} \mathbf{W}_{b}(t_{j}),$$

where S_1, \ldots, S_p are independent centered Gaussian random variables with variance

$$\operatorname{var}(S_b) = \sum_{j=1}^q \sum_{j'=1}^q \lambda_{b,j} \lambda_{b,j'} \Sigma_{j,j'}$$

As a consequence, S is a centered Gaussian random variable with variance

$$\operatorname{var}(S) = \sum_{b=1}^{p} \operatorname{var}(S_{b}) = \sum_{b=1}^{p} \sum_{b'=1}^{p} \sum_{j=1}^{q} \sum_{j'=1}^{q} \lambda_{b,j} \lambda_{b,j'} \Sigma_{j,j'} \times (\mathbf{I}_{p})_{b,b'}$$

As a conclusion, $\operatorname{vec}(\mathbf{W}^{\star}) \sim \mathcal{N}_{pq}(\mathbf{0}, \boldsymbol{\Sigma} \otimes \mathbf{I}_p)$ and thus $\mathbf{W}^{\star} \sim \mathcal{M}\mathcal{N}_{p,q}(\mathbf{0}, \boldsymbol{\Sigma}, \mathbf{I}_p)$. Finally, $\mathbf{Y}^{\star} = \mathbf{A}\mathbf{W}^{\star} + \mathbf{M} \sim \mathcal{M}\mathcal{N}_{p,q}(\mathbf{M}, \boldsymbol{\Sigma}, \mathbf{A}\mathbf{A}^{\top})$, see (Dawid, 1981, Example 1).

Proof of Lemma 1. Combining (4) and (9) yields that the density of $\mathbf{Y}^{i,\star}$ conditionally to $Z_i = c$ is given for all i = 1, ..., n by

(22)
$$p_{i,c}(\mathbf{y}) = (2\pi)^{-pq_i/2} \det(\mathbf{\Sigma}^{c,i}(\boldsymbol{\theta}_c))^{-p/2} \det(\mathbf{A}_c \mathbf{A}_c^{\top})^{-q_i/2} \\ \times \exp\left(-\frac{1}{2} \operatorname{tr}\left[(\mathbf{A}_c \mathbf{A}_c^{\top})^{-1} (\mathbf{y} - \boldsymbol{\alpha}_c \mathbf{B}^i) (\mathbf{\Sigma}^{c,i}(\boldsymbol{\theta}_c))^{-1} (\mathbf{y} - \boldsymbol{\alpha}_c \mathbf{B}^i)^{\top}\right]\right).$$

The likelihood is thus defined as

$$\prod_{c=1}^{C}\prod_{i|Z_i=c}p_{i,c}(\mathbf{Y}^{i,\star}),$$

and the negative log-likelihood can be written as

$$\mathcal{L} = -\sum_{c=1}^{C} \sum_{i|Z_i=c} \log p_{i,c}(\mathbf{Y}^{i,\star}) := \frac{1}{2} \sum_{c=1}^{C} \ell_c(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c, \mathbf{A}_c \mathbf{A}_c^{\mathsf{T}}) + \frac{p \log(2\pi)}{2} \sum_{c=1}^{C} \sum_{i|Z_i=c} q_i,$$

with, for all $c = 1, \ldots, C$,

$$\ell_{c}(\boldsymbol{\alpha}_{c},\boldsymbol{\theta}_{c},\mathbf{A}_{c}\mathbf{A}_{c}^{\mathsf{T}}) = p \sum_{i|Z_{i}=c} \log \det(\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})) + \sum_{i|Z_{i}=c} q_{i} \log \det(\mathbf{A}_{c}\mathbf{A}_{c}^{\mathsf{T}}) + \sum_{i|Z_{i}=c} \operatorname{tr}\left[(\mathbf{A}_{c}\mathbf{A}_{c}^{\mathsf{T}})^{-1}(\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_{c}\mathbf{B}^{i})(\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c}))^{-1}(\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_{c}\mathbf{B}^{i})^{\mathsf{T}}\right]$$

The conclusion follows.

Proof of Proposition 2. (*i*) Let us first consider the differential of $\ell_c(\alpha_c, \theta_c, \mathbf{A}_c \mathbf{A}_c^{\mathsf{T}})$ w.r.t. α_c :

$$\begin{aligned} \mathrm{d}\ell_{c}(\boldsymbol{\alpha}_{c}) &:= \sum_{i|Z_{i}=c} \mathrm{tr}\left(\{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1}\left[\mathrm{d}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})\{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})^{\top}\right]\right) \\ &= -\sum_{i|Z_{i}=c} \mathrm{tr}\left(\{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1}(\mathrm{d}\boldsymbol{\alpha}_{c})\mathbf{B}^{i}\{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})^{\top}\right) \\ &-\sum_{i|Z_{i}=c} \mathrm{tr}\left(\{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})\{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1}(\mathbf{B}^{i})^{\top}(\mathrm{d}\boldsymbol{\alpha}_{c})^{\top}\right) \\ &= -2\sum_{i|Z_{i}=c} \mathrm{tr}\left(\{\mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1}(\mathbf{Y}^{i,\star}-\boldsymbol{\alpha}_{c}\mathbf{B}^{i})\{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1}(\mathbf{B}^{i})^{\top}(\mathrm{d}\boldsymbol{\alpha}_{c})^{\top}\right),\end{aligned}$$

by remarking that both terms are equal in view of the properties of the trace operator. Moreover, from Kronecker product properties (Schott, 2016, Theorem 8.12), one has

$$d\ell_c(\boldsymbol{\alpha}_c) = -2\sum_{i|Z_i=c} \operatorname{vec}(d\boldsymbol{\alpha}_c)^\top \left(\mathbf{B}^i \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)\}^{-1} \otimes \{\mathbf{A}_c \mathbf{A}_c^\top\}^{-1}\right) \operatorname{vec}(\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i)$$
$$= -2(\operatorname{dvec}(\boldsymbol{\alpha}_c))^\top \operatorname{vec}\left(\{\mathbf{A}_c \mathbf{A}_c^\top\}^{-1} \sum_{i|Z_i=c} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)\}^{-1} (\mathbf{B}^i)^\top\right).$$

Interpreting the above result as a scalar product and using the "broad" definition of matrix derivative defined in (Magnus, 2010), if follows:

$$\frac{\partial \ell_c(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c, \mathbf{A}_c \mathbf{A}_c^{\top})}{\partial \boldsymbol{\alpha}_c} = -2 \operatorname{vec} \left(\{ \mathbf{A}_c \mathbf{A}_c^{\top} \}^{-1} \sum_{i | Z_i = c} (\mathbf{Y}^{i, \star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{ \mathbf{\Sigma}^{c, i}(\boldsymbol{\theta}_c) \}^{-1} (\mathbf{B}^i)^{\top} \right).$$

Setting this partial derivative to zero yields

$$\sum_{i|Z_i=c} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} (\mathbf{B}^i)^\top = 0,$$

or equivalently,

$$\boldsymbol{\alpha}_{c} = \left[\sum_{i|Z_{i}=c} \mathbf{Y}^{i,\star} \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1} (\mathbf{B}^{i})^{\top}\right] \left[\sum_{i|Z_{i}=c} \mathbf{B}^{i} \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1} (\mathbf{B}^{i})^{\top}\right]^{-1},$$

which is the desired result. Second, let us consider the differential of $\ell_c(\alpha_c, \theta_c, \mathbf{A}_c\mathbf{A}_c^{\mathsf{T}})$ w.r.t. $\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}$:

$$d\ell_c(\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}) = Q_c d\log \det(\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}) + dtr(\mathbf{N}(\boldsymbol{\theta}_c)\{\mathbf{A}_c\mathbf{A}_c^{\mathsf{T}}\}^{-1}),$$

where $\mathbf{N}(\boldsymbol{\theta}_c) = \sum_{i|Z_i=c} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i)^{\top}$. From (Schott, 2016, Example 9.6), the associated partial derivative vanishes for

$$\mathbf{A}_{c}\mathbf{A}_{c}^{\mathsf{T}} = \frac{\mathbf{N}(\boldsymbol{\theta}_{c})}{Q_{c}} = \frac{1}{Q_{c}}\sum_{i|Z_{i}=c} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_{c}\mathbf{B}^{i}) \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_{c})\}^{-1} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_{c}\mathbf{B}^{i})^{\mathsf{T}},$$

and the result is proved.

(*ii*) Introduce $\boldsymbol{\beta}^{c,i}(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c) = (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)\}^{-1}$ and consider the *k*th coordinate of the gradient of $\ell_c(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c, \mathbf{A}_c \mathbf{A}_c^{\mathsf{T}})$ w.r.t. $\boldsymbol{\theta}$:

$$\begin{aligned} \frac{\partial \ell_c(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c, \mathbf{A}_c \mathbf{A}_c^{\top})}{\partial \theta_k} &= p \sum_{i|Z_i=c} \frac{\partial}{\partial \theta_k} \log \det(\boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)) + \frac{\partial}{\partial \theta_k} \operatorname{tr} \left(\mathbf{N}(\boldsymbol{\theta}_c) \{\mathbf{A}_c \mathbf{A}_c^{\top}\}^{-1} \right) \\ &= p \sum_{i|Z_i=c} \operatorname{tr} \left(\{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} \frac{\partial \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)}{\partial \theta_k} \right) \\ &- \sum_{i|Z_i=c} \operatorname{tr} \left((\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i) \{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} \frac{\partial \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)}{\partial \theta_k} \{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} (\mathbf{Y}^{i,\star} - \boldsymbol{\alpha}_c \mathbf{B}^i)^{\top} \{\mathbf{A}_c \mathbf{A}_c^{\top} \}^{-1} \right) \\ &= p \sum_{i|Z_i=c} \operatorname{tr} \left(\{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} \frac{\partial \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)}{\partial \theta_k} \right) \\ &- \sum_{i|Z_i=c} \operatorname{tr} \left(\{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} \frac{\partial \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c)}{\partial \theta_k} \right) \\ &= \sum_{i|Z_i=c} \operatorname{tr} \left(p \{ \boldsymbol{\Sigma}^{c,i}(\boldsymbol{\theta}_c) \}^{-1} - \boldsymbol{\beta}^{c,i}(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c)^{\top} \{\mathbf{A}_c \mathbf{A}_c^{\top} \}^{-1} \boldsymbol{\beta}^{c,i}(\boldsymbol{\alpha}_c, \boldsymbol{\theta}_c) \right] \frac{\partial \boldsymbol{\Sigma}^{c,i}}{\partial \theta_k} (\boldsymbol{\theta}_c) \right). \end{aligned}$$

The result is proved.

Proof of Proposition 3. Let $\mathbf{Y}^{\star} = (\mathbf{Y}(t_1), \dots, \mathbf{Y}(t_q))$ be a $p \times q$ random matrix where, conditionally to Z = c, $\mathbf{Y} \sim \mathcal{MGP}_c(\alpha_c \mathbf{b}, K_c, \mathbf{A}_c)$. Recall that Proposition 1 yields $\operatorname{vec}(\mathbf{Y}^{\star}) \sim \mathcal{N}_{pq}(\operatorname{vec}(\alpha_c \mathbf{B}), \Sigma^c(\theta_c) \otimes \mathbf{A}_c \mathbf{A}_c^{\top})$, where $\Sigma^c(\theta_c)$ is defined for all $(j, j') \in \{1, \dots, q\}^2$ by $\Sigma^c(\theta_c)_{j,j'} = K_c(t_j, t_{j'}|\theta_c)$ and $\mathbf{B} = (\mathbf{b}(t_1), \dots, \mathbf{b}(t_q))$ is a $J \times q$ design matrix. Let $t^{\dagger} \in \mathcal{T}$ be an unobserved time, *i.e.* $t^{\dagger} \neq t_k$, for all $k \in \{1, \dots, q\}$, and $\mathbf{k}_c(t^{\dagger}) = (K_c(t^{\dagger}, t_1|\theta_c), \dots, K_c(t^{\dagger}, t_q|\theta_c))^{\top}$. Then, classical properties on conditional Gaussian random vectors (see for instance (Bilodeau and Brenner, 2008, p. 63)) entail that, conditionally to Z = c and $\operatorname{vec}(\mathbf{Y}^{\star}) = \operatorname{vec}(\mathbf{y}^{\star}), \mathbf{Y}(t^{\dagger})$ follows the *p*-variate Gaussian distribution $\mathcal{N}_p(\boldsymbol{\mu}_c(t^{\dagger}, \mathbf{y}^{\star}), \mathbf{\Lambda}_c(t^{\dagger}))$ with, on the one hand

$$\begin{split} \boldsymbol{\mu}_{c}(t^{\dagger},\mathbf{y}^{\star}) &= \boldsymbol{\alpha}_{c}\mathbf{b}(t^{\dagger}) + [\mathbf{k}_{c}(t^{\dagger})^{\top} \otimes \mathbf{A}_{c}\mathbf{A}_{c}^{\top}] \{\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c}) \otimes \mathbf{A}_{c}\mathbf{A}_{c}^{\top}\}^{-1} \operatorname{vec}(\mathbf{y}^{\star} - \boldsymbol{\alpha}_{c}\mathbf{B}) \\ &= \boldsymbol{\alpha}_{c}\mathbf{b}(t^{\dagger}) + [\mathbf{k}_{c}(t^{\dagger})^{\top} \otimes \mathbf{A}_{c}\mathbf{A}_{c}^{\top}] \{\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1} \otimes (\mathbf{A}_{c}\mathbf{A}_{c}^{\top})^{-1}\} \operatorname{vec}(\mathbf{y}^{\star} - \boldsymbol{\alpha}_{c}\mathbf{B}) \\ &= \boldsymbol{\alpha}_{c}\mathbf{b}(t^{\dagger}) + \left[\{\mathbf{k}_{c}(t^{\dagger})^{\top}\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1}\} \otimes \{(\mathbf{A}_{c}\mathbf{A}_{c}^{\top})(\mathbf{A}_{c}\mathbf{A}_{c}^{\top})^{-1}\} \right] \operatorname{vec}(\mathbf{y}^{\star} - \boldsymbol{\alpha}_{c}\mathbf{B}) \\ &= \boldsymbol{\alpha}_{c}\mathbf{b}(t^{\dagger}) + \left[\{\mathbf{k}_{c}(t^{\dagger})^{\top}\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1}\} \otimes \mathbf{I}_{p} \right] \operatorname{vec}(\mathbf{y}^{\star} - \boldsymbol{\alpha}_{c}\mathbf{B}) \\ &= \boldsymbol{\alpha}_{c}\mathbf{b}(t^{\dagger}) + \operatorname{vec}\left(\mathbf{I}_{p}(\mathbf{y}^{\star} - \hat{\boldsymbol{\alpha}}_{c}\mathbf{B})\{\mathbf{k}_{c}(t^{\dagger})^{\top}\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1}\}^{\top} \right) \\ &= \boldsymbol{\alpha}_{c}\mathbf{b}(t^{\dagger}) + (\mathbf{y}^{\star} - \boldsymbol{\alpha}_{c}\mathbf{B})\boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1}\mathbf{k}_{c}(t^{\dagger}), \end{split}$$

and on the other hand,

$$\begin{split} \mathbf{\Lambda}_{c}(t^{\dagger}) &= K_{c}(t^{\dagger}, t^{\dagger} | \boldsymbol{\theta}_{c}) \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top} - [\mathbf{k}_{c}(t^{\dagger})^{\top} \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top}] \left\{ \boldsymbol{\Sigma}^{c} \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top}(\boldsymbol{\theta}_{c}) \right\}^{-1} [\mathbf{k}_{c}(t^{\dagger}) \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top}] \\ &= K_{c}(t^{\dagger}, t^{\dagger} | \boldsymbol{\theta}_{c}) \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top} - \left[(\mathbf{k}_{c}(t^{\dagger})^{\top} \boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1}) \otimes \mathbf{I}_{p} \right] [\mathbf{k}_{c}(t^{\dagger}) \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top}] \\ &= K_{c}(t^{\dagger}, t^{\dagger} | \boldsymbol{\theta}_{c}) \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top} - \left[(\mathbf{k}_{c}(t^{\dagger})^{\top} \boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1} \mathbf{k}_{c}(t^{\dagger}) \right] \otimes (\mathbf{I}_{p} \mathbf{A}_{c} \mathbf{A}_{c}^{\top}) \\ &= \left[K_{c}(t^{\dagger}, t^{\dagger} | \boldsymbol{\theta}_{c}) - \mathbf{k}_{c}(t^{\dagger})^{\top} \boldsymbol{\Sigma}^{c}(\boldsymbol{\theta}_{c})^{-1} \mathbf{k}_{c}(t^{\dagger}) \right] \otimes \mathbf{A}_{c} \mathbf{A}_{c}^{\top}. \end{split}$$

The result is proved.

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