



Brief paper

Geometric properties of partial least squares for process monitoring[☆]Gang Li^a, S. Joe Qin^{b,c,*}, Donghua Zhou^{a,*}^a Department of Automation, TNList, Tsinghua University, Beijing 100084, PR China^b The Mork Family Department of Chemical Engineering and Materials Science, University of Southern California, Los Angeles, CA 90089, USA^c Ming Hsieh Department of Electrical Engineering, University of Southern California, Los Angeles, CA 90089, USA

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ABSTRACT

Projection to latent structures or partial least squares (PLS) produces output-supervised decomposition on input \mathbf{X} , while principal component analysis (PCA) produces unsupervised decomposition of input \mathbf{X} . In this paper, the effect of output \mathbf{Y} on the \mathbf{X} -space decomposition in PLS is analyzed and geometric properties of the PLS structure are revealed. Several PLS algorithms are compared in a geometric way for the purpose of process monitoring. A numerical example and a case study are given to illustrate the analysis results.

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1. Introduction

Multivariate statistical process monitoring (SPM) has been successfully used in the monitoring of different industrial processes over the past two decades, including chemicals, polymers, and microelectronics manufacturing. Multivariate statistical process control charts based on principal component analysis (PCA), projection to latent structures (PLS), and other data-based structures have received great success in practice (Kresta, MacGregor & Marlin, 1991; Qin, 2003; Wise & Gallagher, 1996; Xia, Howell & Thornhill, 2005). Besides, fault reconstruction and estimation can be performed based on the latent space of PCA (Dunia & Qin, 1998; Qin, 2003), which enhances the application of SPM significantly.

To monitor all the variations and abnormal situations of input measurements (\mathbf{X}), one can perform a PCA decomposition on the \mathbf{X} -space. However, a more important objective of process monitoring is to provide assurance of good product quality that is impacted by the process conditions. Quality variables (\mathbf{Y}) are affected by the processing conditions reflected in the measured \mathbf{X} -data and possibly additional unmeasured factors. The quality data \mathbf{Y} are often difficult to measure, and often come very infrequently with significant measurement delays. To monitor variations in the process

variables that are most relevant to quality variables (\mathbf{Y}), one can perform PLS decomposition on \mathbf{X} -space.

The basic concepts and algorithms of PLS can be found in the chemometrics literature (Di Ruscio, 2000; Hóskuldsson, 1988; Ter Braak & De Jong, 1998). PLS has been used in multivariate monitoring of process operating performance, which is almost exactly in the same way as PCA-based monitoring (Kresta et al., 1991). Several variants of PLS have been proposed for monitoring, such as multi-block PLS (MacGregor, Jaeckle, Kiparissides & Koutoudi, 1994), dynamic PLS (Lee, Han & Yoon, 2004), recursive PLS (Qin, 1998) and multi-phase PLS (Lu, Gao & Wang, 2004).

Although PLS-based monitoring has been used for a long time, the property of the latent space induced by PLS has not been analyzed for process monitoring. While PCA-based monitoring methods are well understood (e.g., Alcalá & Qin, 2009), it is not clear how \mathbf{Y} affects the decomposition of \mathbf{X} -space and the outcome of process monitoring. Westerhuis, Gurden and Smilde (2000) proposed generalized T^2 and Q statistics for many latent variable models (including PLS). The structure they used can be regarded as the structure of simplified PLS (SIMPLS) (De Jong, 1993). However, it is still an open question as to which PLS algorithm is the most appropriate for process monitoring.

In this paper, we show geometrically the \mathbf{X} -space decomposition supervised by \mathbf{Y} using PLS relative to PCA. Then, we reveal the geometric property of the decomposition induced by PLS. It is made clear why and how PLS should be used in process monitoring. Three PLS algorithms are analyzed in detail and the most appropriate structure for monitoring is pointed out.

The remainder of this paper is organized as follows. Section 2 reviews the standard PLS algorithm and its properties. Then, we

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Table 1
X-deflated NIPALS algorithm (Dayal & MacGregor, 1997).

<p>Center the columns of \mathbf{X}, \mathbf{Y} to zero mean and scale them to unit variance. Set $i = 1$ and $\mathbf{X}_1 = \mathbf{X}$.</p> <ol style="list-style-type: none"> 1. Set \mathbf{u}_i equal to any column of \mathbf{Y}. 2. $\mathbf{w}_i = \mathbf{X}_i^T \mathbf{u}_i / \ \mathbf{X}_i^T \mathbf{u}_i\$. 3. $\mathbf{t}_i = \mathbf{X}_i \mathbf{w}_i$. 4. $\mathbf{q}_i = \mathbf{Y}^T \mathbf{t}_i / \mathbf{t}_i^T \mathbf{t}_i$. 5. $\mathbf{u}_i = \mathbf{Y} \mathbf{q}_i$. <p>If \mathbf{t}_i converges, go to Step 6, else return to Step 2.</p> <ol style="list-style-type: none"> 6. $\mathbf{p}_i = \mathbf{X}_i^T \mathbf{t}_i / \mathbf{t}_i^T \mathbf{t}_i$. 7. $\mathbf{X}_{i+1} = \mathbf{X}_i - \mathbf{t}_i \mathbf{p}_i^T$. <p>Set $i = i + 1$ and return to step 1. Terminate if $i > A$.</p>

discuss the effect of \mathbf{Y} on the \mathbf{X} -space decomposition in Section 3. The geometric properties of PLS on \mathbf{X} -space decomposition are discussed in Section 4. Other PLS variants are analyzed in a similar way. Following that, we discuss the monitoring problem using PLS and its variants in Section 5. Section 6 uses a numerical example and a case study to illustrate the analysis results. Finally, we present conclusions in the last section.

2. Projection to latent spaces (PLS)

Given an input matrix $\mathbf{X} \in \mathbb{R}^{n \times m}$ consisting of n samples with m process variables per sample, and an output matrix $\mathbf{Y} \in \mathbb{R}^{n \times p}$ with p quality variables per sample, PLS projects (\mathbf{X}, \mathbf{Y}) to a low-dimensional space defined by a small number of latent variables $(\mathbf{t}_1, \dots, \mathbf{t}_A)$ (A is the PLS component number) as follows:

$$\begin{cases} \mathbf{X} = \mathbf{T}\mathbf{P}^T + \mathbf{E} \\ \mathbf{Y} = \mathbf{T}\mathbf{Q}^T + \mathbf{F} \end{cases} \quad (1)$$

where $\mathbf{T} = [\mathbf{t}_1, \dots, \mathbf{t}_A]$ is the score matrix, $\mathbf{P} = [\mathbf{p}_1, \dots, \mathbf{p}_A]$ is the loading matrix for \mathbf{X} and $\mathbf{Q} = [\mathbf{q}_1, \dots, \mathbf{q}_A]$ is the loading matrix for \mathbf{Y} . \mathbf{E} and \mathbf{F} are the modeling residual of \mathbf{X} and \mathbf{Y} . The data matrices \mathbf{X} , \mathbf{Y} are usually scaled to unit variance and zero mean. A nonlinear iterative partial least-squares algorithm (NIPALS) to perform PLS is described in Table 1. The objective of PLS embedded in this algorithm is to find the solution of the following problem:

$$\max_{\mathbf{w}_i, \mathbf{q}_i} \mathbf{w}_i^T \mathbf{X}_i^T \mathbf{Y}_i \mathbf{q}_i$$

s.t. $\|\mathbf{w}_i\| = 1, \|\mathbf{q}_i\| = 1$

where $\mathbf{w}_i, \mathbf{q}_i$ are weight vectors that yield $\mathbf{t}_i = \mathbf{X}_i \mathbf{w}_i$ and $\mathbf{u}_i = \mathbf{Y}_i \mathbf{q}_i$, respectively. Denoting $\mathbf{W} = [\mathbf{w}_1, \dots, \mathbf{w}_A]$, \mathbf{T} cannot be calculated from \mathbf{X} directly using \mathbf{W} . Let

$$\mathbf{r}_1 = \mathbf{w}_1, \quad \mathbf{r}_i = \prod_{j=1}^{i-1} (\mathbf{I}_m - \mathbf{w}_j \mathbf{p}_j^T) \mathbf{w}_i, \quad i > 1 \quad (2)$$

and $\mathbf{R} = [\mathbf{r}_1, \dots, \mathbf{r}_A]$. Then, the score matrix \mathbf{T} can be computed from the original \mathbf{X} as follows:

$$\mathbf{T} = \mathbf{X}\mathbf{R} \quad (3)$$

\mathbf{P} , \mathbf{R} and \mathbf{W} have the following relationship (Dayal & MacGregor, 1997; De Jong, 1993):

$$\mathbf{R} = \mathbf{W}(\mathbf{P}^T \mathbf{W})^{-1} \quad (4)$$

$$\mathbf{P}^T \mathbf{R} = \mathbf{R}^T \mathbf{P} = \mathbf{W}^T \mathbf{W} = \mathbf{I}_A. \quad (5)$$

In the PLS literature, the case of a single output is referred to as PLS1 and that for multiple outputs is referred to as PLS2. When there are several output variables, performing PLS1 of each output separately does not give the same results as PLS2 of multiple outputs jointly. Since outputs from one process are usually interrelated, it is appropriate to use PLS2 to capture the interrelationship among the output variables.

3. The effect of \mathbf{Y} on the \mathbf{X} -space decomposition

Many researchers use the PCA-based monitoring techniques for PLS decomposition of the \mathbf{X} -space. However, the PLS decomposition can be radically different from the PCA decomposition, which makes one wonder whether the PLS-based monitoring should be different from the PCA-based monitoring techniques. In this section, we demonstrate the impact of \mathbf{Y} on the decomposition of \mathbf{X} -space in general, and then visualize the result geometrically.

Suppose \mathbf{X} has the following PCA decomposition:

$$\mathbf{X} = \mathbf{t}_1 \mathbf{v}_1^T + \dots + \mathbf{t}_l \mathbf{v}_l^T \quad (6)$$

where \mathbf{v}_i ($1 \leq i \leq l$) are the orthonormal eigenvectors related to nonzero eigenvalues of $\mathbf{X}^T \mathbf{X}$, $\lambda_1 \geq \dots \geq \lambda_l > 0$ and $l = \text{rank}(\mathbf{X}) \leq m$. In PCA \mathbf{v}_i ($1 \leq i \leq l$) alone define the decomposition of the input space. In PLS however, the input space decomposition is defined by two matrices, \mathbf{P} and \mathbf{R} . Therefore, the angle between \mathbf{r}_i and \mathbf{p}_i , unless it is zero, reflects the impact of \mathbf{Y} on the decomposition of \mathbf{X} -space in PLS. For the ease of presentation, we drop the subscript i for the moment.

The PLS weight vector \mathbf{r} is in $\text{Span}\{\mathbf{v}_1, \dots, \mathbf{v}_l\}$ according to the properties of PLS. Therefore,

$$\mathbf{r} = r \sum_{i=1}^l \alpha_i \mathbf{v}_i \quad (7)$$

where $r = \|\mathbf{r}\|$ and

$$\sum_{i=1}^l \alpha_i^2 = 1. \quad (8)$$

Then,

$$\mathbf{p} = \mathbf{X}^T \mathbf{t} / \mathbf{t}^T \mathbf{t} = \frac{\mathbf{X}^T \mathbf{X} \mathbf{r}}{\mathbf{t}^T \mathbf{t}} = \frac{\sum_{i=1}^l \lambda_i \alpha_i \mathbf{v}_i}{r \sum_{i=1}^l \lambda_i \alpha_i^2}. \quad (9)$$

From (5), we have $\mathbf{r}^T \mathbf{p} = 1$ for each dimension. Therefore,

$$\cos \angle(\mathbf{r}, \mathbf{p}) = \frac{1}{r \|\mathbf{p}\|} = \frac{\sum_{i=1}^l \lambda_i \alpha_i^2}{\sqrt{\sum_{i=1}^l \lambda_i^2 \alpha_i^2}} \quad (10)$$

and

$$\max \angle(\mathbf{r}, \mathbf{p}) = \arccos \frac{2\sqrt{\lambda_1 \lambda_2}}{\lambda_1 + \lambda_2}. \quad (11)$$

Result (11) is obtained by minimizing (10) subject to (8). The solution process is omitted due to page limitation.

Several points can be summarized about the effect of \mathbf{Y} on PLS decomposition of the \mathbf{X} -space.

- Generally, \mathbf{r} and \mathbf{p} in PLS have a nonzero angle.
- $\angle(\mathbf{r}, \mathbf{p})$ has an upper bound determined by (11). This upper bound increases with the difference among λ_i .
- If \mathbf{r} is an eigenvector of $\mathbf{X}^T \mathbf{X}$, all but one α_i is zero, which makes (10) equal to one and $\angle(\mathbf{r}, \mathbf{p}) = 0$.
- If λ_i are equal, $\angle(\mathbf{r}, \mathbf{p}) = 0$.

To visualize the results geometrically, consider the special case of two inputs and one output. Suppose $\mathbf{X} = [\mathbf{x}_1, \mathbf{x}_2] = \mathbf{t}_1 \mathbf{v}_1^T + \mathbf{t}_2 \mathbf{v}_2^T \in \mathbb{R}^{n \times 2}$, $\mathbf{y} = c_1 \mathbf{t}_1 + c_2 \mathbf{t}_2 \in \mathbb{R}^{n \times 1}$. Then, (11) reduces to

$$\max \angle(\mathbf{r}, \mathbf{p}) = \arccos \frac{2\sqrt{\lambda_1 \lambda_2}}{\lambda_1 + \lambda_2}. \quad (12)$$

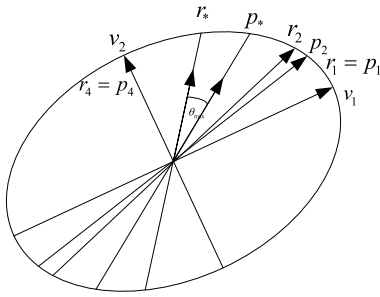


Fig. 1. Effect on X-space decomposition by \mathbf{y} .

From (12), it can be observed that $\max \angle(\mathbf{r}, \mathbf{p})$ is close to 0 when λ_2 is close to λ_1 .

Fig. 1 describes different cases in a geometric way. In PCA, the X-space is decomposed by \mathbf{v}_1 and \mathbf{v}_2 . In PLS, it depends on \mathbf{X} and \mathbf{y} jointly. If $c_2 = 0$, then \mathbf{r} coincide with \mathbf{v}_1 , which forms the same decomposition as PCA. \mathbf{r}_1 and \mathbf{p}_1 in Fig. 1 denote this case. If \mathbf{y} is more related to \mathbf{t}_2 and less to \mathbf{t}_1 , then \mathbf{r} is chosen to be farther from \mathbf{v}_1 and closer to \mathbf{v}_2 , which causes $\angle(\mathbf{r}, \mathbf{p})$ to increase. The case is described by $\mathbf{r}_2, \mathbf{p}_2$ in Fig. 1. The largest angle between vectors \mathbf{r} and \mathbf{p} are represented by $\mathbf{r}^*, \mathbf{p}^*$. If $c_1 = 0$, \mathbf{y} is only related to \mathbf{t}_2 , then \mathbf{r} and \mathbf{p} overlap again ($\mathbf{r} = \mathbf{p} = \mathbf{v}_2$), which are denoted by \mathbf{r}_4 and \mathbf{p}_4 , respectively.

Therefore, the impact of \mathbf{Y} on the decomposition of the X-space depends on how ellipsoidal the covariance of \mathbf{X} is (i.e. unequal λ_i) and how much \mathbf{Y} is correlated to the leading PCA scores of \mathbf{X} . In typical engineering applications, \mathbf{X} data are highly correlated. Therefore, the covariance of \mathbf{X} is usually very ellipsoidal. As a consequence, the upper bound of $\angle(\mathbf{p}, \mathbf{r})$ can be significant. If \mathbf{Y} is mostly correlated to the leading PCA scores of \mathbf{X} , the PLS decomposition of the X-space is similar to the PCA decomposition of the X-space. In this case, the PCA monitoring techniques can be directly applied to the PLS model and residuals. If, on the other hand, \mathbf{Y} is highly correlated to the non-leading PCA scores of \mathbf{X} , the PLS decomposition of the X-space in (1) can be very different from the PCA decompositions of the X-space. The variance left in the X-residual (\mathbf{E}) can be large and the direct application of Q statistic monitoring on the X-residual is questionable.

4. Geometric properties of PLS

4.1. Space decomposition of PCA

In PCA the input vector \mathbf{x} is decomposed as follows (Qin, 2003):

$$\begin{aligned} \mathbf{x} &= \hat{\mathbf{x}} + \tilde{\mathbf{x}} \\ \hat{\mathbf{x}} &= \mathbf{P}\mathbf{P}^T\mathbf{x} \in S_p \equiv \text{Span}\{\mathbf{P}\} \\ \tilde{\mathbf{x}} &= (\mathbf{I} - \mathbf{P}\mathbf{P}^T)\mathbf{x} \in S_r \equiv \text{Span}\{\mathbf{P}\}^\perp \end{aligned} \quad (13)$$

where \mathbf{P} is the loading matrix of PCA, and $\text{Span}\{\mathbf{P}\}^\perp$ is the orthogonal complement of $\text{Span}\{\mathbf{P}\}$. $\mathbf{P}\mathbf{P}^T$ and $\mathbf{I} - \mathbf{P}\mathbf{P}^T$ are both orthogonal projectors. Thus, (13) is an orthogonal decomposition.

4.2. Space decomposition of PLS

From (5), we can easily obtain

$$(\mathbf{P}\mathbf{R}^T)^2 = \mathbf{P}\mathbf{R}^T. \quad (14)$$

Thus, $\mathbf{P}\mathbf{R}^T$ is an idempotent matrix. Generally speaking, $\mathbf{P}\mathbf{R}^T$ is not symmetric. Therefore, $\mathbf{P}\mathbf{R}^T$ is an oblique projector (Zhang, 2004).

Lemma 1. Let $\Pi_{P|R^\perp}$ denote the projector onto the subspace $\text{Span}\{\mathbf{P}\}$, along the subspace $\text{Span}\{\mathbf{R}\}^\perp$.

$$\begin{aligned} \Pi_{P|R^\perp} &= \mathbf{P}\mathbf{R}^T \\ \Pi_{R^\perp|P} &= \mathbf{I} - \mathbf{P}\mathbf{R}^T. \end{aligned} \quad (15)$$

The proof is given in Appendix A.

From Lemma 1, we have the following theorem on the PLS decomposition:

Theorem 1. PLS induces an oblique decomposition on input variable space:

$$\begin{aligned} \mathbf{x} &= \hat{\mathbf{x}} + \tilde{\mathbf{x}} \\ \hat{\mathbf{x}} &= \mathbf{P}\mathbf{R}^T\mathbf{x} \in S_p \equiv \text{Span}\{\mathbf{P}\} \\ \tilde{\mathbf{x}} &= (\mathbf{I} - \mathbf{P}\mathbf{R}^T)\mathbf{x} \in S_r \equiv \text{Span}\{\mathbf{R}\}^\perp. \end{aligned} \quad (16)$$

The theorem can be proven by noting from Lemma 1 that

$$\text{Span}\{\mathbf{I} - \mathbf{P}\mathbf{R}^T\} = \text{Span}\{\mathbf{R}\}^\perp. \quad (17)$$

Different from PCA, $\hat{\mathbf{x}}$ is not orthogonal to $\tilde{\mathbf{x}}$ in PLS. Therefore, we conclude that in the PLS model, $\hat{\mathbf{x}}$ is the projection of \mathbf{x} onto $\text{Span}\{\mathbf{P}\}$ along $\text{Span}\{\mathbf{R}\}^\perp$ and $\tilde{\mathbf{x}}$ is the projection of \mathbf{x} onto $\text{Span}\{\mathbf{R}\}^\perp$ along $\text{Span}\{\mathbf{P}\}$.

4.3. Variants of PLS

The weight-deflated PLS (W-PLS) is a variant of PLS proposed by Helland (1988), which has the same prediction ability as the standard PLS. The relationships between W-PLS and PLS are as follows (Helland, 1988).

1. \mathbf{W} in W-PLS is identical to \mathbf{W} in the standard PLS.
2. $\text{Span}\{\mathbf{T}\}$ in W-PLS is the same as $\text{Span}\{\mathbf{T}\}$ in the standard PLS.

From (4), we know that \mathbf{W} and \mathbf{R} share the same column space, which means

$$\text{Span}\{\mathbf{W}\} = \text{Span}\{\mathbf{R}\}. \quad (18)$$

W-PLS induces the following input space decomposition,

$$\begin{aligned} \mathbf{x} &= \hat{\mathbf{x}}_w + \tilde{\mathbf{x}}_w \\ \hat{\mathbf{x}}_w &= \mathbf{W}\mathbf{W}^T\mathbf{x} \in \text{Span}\{\mathbf{R}\} \\ \tilde{\mathbf{x}}_w &= (\mathbf{I} - \mathbf{W}\mathbf{W}^T)\mathbf{x} \in \text{Span}\{\mathbf{R}\}^\perp \end{aligned} \quad (19)$$

which is an orthogonal decomposition.

De Jong (1993) proposed SIMPLS as an alternative approach of PLS. SIMPLS is proven to be identical to PLS1 for one output, and different from PLS2 with multiple outputs (Ter Braak & De Jong, 1998).

A difference between PLS and SIMPLS is in which subspace to search for \mathbf{w}_i and how to obtain \mathbf{r}_i . In PLS, \mathbf{w}_i is found in $\text{Span}\{\mathbf{R}_{i-1}\}^\perp$, and \mathbf{r}_i is obtained by projecting \mathbf{w}_i to $\text{Span}\{\mathbf{P}_{i-1}\}^\perp$ along $\text{span}\{\mathbf{R}_{i-1}\}$. In SIMPLS, \mathbf{w}_i is found in $\text{Span}\{\mathbf{P}_{i-1}\}^\perp$ directly and $\mathbf{r}_i = \mathbf{w}_i$, because $\mathbf{w}_i \in \text{Span}\{\mathbf{P}_{i-1}\}^\perp$.

SIMPLS provides another decomposition of X-space as follows:

$$\begin{aligned} \mathbf{x} &= \hat{\mathbf{x}}_s + \tilde{\mathbf{x}}_s \\ \hat{\mathbf{x}}_s &= \mathbf{P}\mathbf{P}^+\mathbf{x} \in \text{Span}\{\mathbf{P}\} \\ \tilde{\mathbf{x}}_s &= (\mathbf{I} - \mathbf{P}\mathbf{P}^+)\mathbf{x} \in \text{Span}\{\mathbf{P}\}^\perp \end{aligned} \quad (20)$$

where $\mathbf{P}^+ = (\mathbf{P}^T\mathbf{P})^{-1}\mathbf{P}^T$ is the generalized inverse of \mathbf{P} . $\mathbf{P}\mathbf{P}^+$ is the orthogonal projector on $\text{Span}\{\mathbf{P}\}$, while $\mathbf{I} - \mathbf{P}\mathbf{P}^+$ is the orthogonal projector on $\text{Span}\{\mathbf{P}\}^\perp$. Therefore, (20) is an orthogonal decomposition of the X-space.

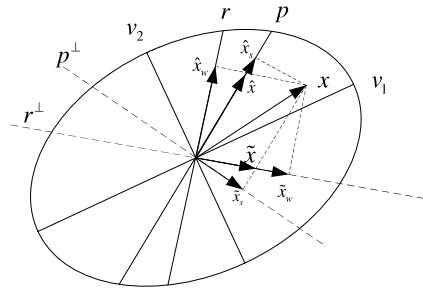


Fig. 2. Decompositions of PLS, W-PLS and SIMPLS.

4.4. Comparison of PLS, W-PLS, and SIMPLS

From (16), (19) and (20), it can be observed that W-PLS, SIMPLS and PLS are different in the decomposition of the \mathbf{X} -space. We depict the decomposition results of PLS, W-PLS and SIMPLS in Fig. 2, where $\hat{\mathbf{x}}$ and $\tilde{\mathbf{x}}$ are from the standard PLS, while $\hat{\mathbf{x}}_w$ and $\tilde{\mathbf{x}}_w$ are from W-PLS, and $\hat{\mathbf{x}}_s$, $\tilde{\mathbf{x}}_s$ are from SIMPLS.

In PCA, both of the following relations hold:

$$\mathbf{T}^T \tilde{\mathbf{X}} = 0 \quad (21)$$

$$\hat{\mathbf{x}}^T \tilde{\mathbf{x}} = 0. \quad (22)$$

Relation (21) describes the orthogonality in the sample space, which guarantees that the scores extracted from \mathbf{X} are uncorrelated to the residuals. Relation (22) describes the orthogonal decomposition in the variable space, which guarantees that $\hat{\mathbf{x}}$ and $\tilde{\mathbf{x}}$ are mutually exclusive. However, in the above PLS algorithms, only one of relations can be guaranteed at the same time. With an oblique projection, the scores are uncorrelated to the residuals but $\hat{\mathbf{x}}$ and $\tilde{\mathbf{x}}$ in the variable space are not orthogonal. With an orthogonal projection, $\hat{\mathbf{x}}$ and $\tilde{\mathbf{x}}$ are orthogonal, but the scores are correlated to residuals. In the next section, we reveal which PLS projection is the most appropriate for process monitoring.

5. Process monitoring with PLS

5.1. Monitoring with PLS

When a fault occurs, one or more measurement variables are affected, which breaks the normal relationship among these variables. PLS decomposes the measured data \mathbf{X} into S_r and S_p subspaces, and fault detection is done by monitoring these subspaces.

If an abnormal situation affects quality \mathbf{Y} , it is thought to happen in the S_p subspace (MacGregor et al., 1994). Otherwise, if a fault has no impact on quality \mathbf{Y} , it is thought to happen in the S_r subspace. T^2 statistic is used to monitor faults in S_p , while Q statistic is used to monitor faults in S_r . Therefore, PLS-based fault detection can tell whether a fault is related to the quality \mathbf{Y} .

Given a new sample \mathbf{x}_{new} , the PLS score and residual can be calculated from (3) and (16):

$$\begin{aligned} \mathbf{t}_{new} &= \mathbf{R}^T \mathbf{x}_{new} \\ \tilde{\mathbf{x}}_{new} &= (\mathbf{I} - \mathbf{P}\mathbf{R}^T) \mathbf{x}_{new}. \end{aligned} \quad (23)$$

Then, two statistics T^2 and Q are defined as follows:

$$\begin{aligned} T^2 &= \mathbf{t}_{new}^T \mathbf{A}^{-1} \mathbf{t}_{new} \\ Q &= \|\tilde{\mathbf{x}}_{new}\|^2 \end{aligned} \quad (24)$$

where $\mathbf{A} = \frac{1}{n-1} \mathbf{T}^T \mathbf{T}$ is the sample covariance matrix of \mathbf{t} , A is the number of PLS components, n is the number of training samples. The control limits for these statistics are given in, for example, MacGregor et al. (1994) and Choi and Lee (2005). If T^2 and Q are within the control limits, the process is normal. If T^2 exceeds the control limit, it is thought that there is a fault related to \mathbf{Y} in the process. If Q exceeds the control limit, it is thought that the fault is unrelated to \mathbf{Y} .

5.2. Monitoring with SIMPLS

SIMPLS calculates the scores and residuals as follows:

$$\begin{aligned} \mathbf{t}_{s,new} &= \mathbf{P}^+ \mathbf{x}_{new} \\ \tilde{\mathbf{x}}_{s,new} &= (\mathbf{I} - \mathbf{P}\mathbf{P}^+) \mathbf{x}_{new}, \end{aligned} \quad (25)$$

which has been used by Westerhuis et al. (2000). Statistics T_s^2 and Q_s are defined similar to (24).

5.2.1. Relation between \mathbf{t} and \mathbf{t}_s

In order to describe the relation between them, the following lemma is introduced.

Lemma 2. Let $D_{\hat{\mathbf{x}}} = \hat{\mathbf{x}}^T \Gamma_{\hat{\mathbf{x}}}^+ \hat{\mathbf{x}}$, where $\Gamma_{\hat{\mathbf{x}}}^+$ is the generalized inverse of $\frac{1}{n-1} \hat{\mathbf{X}}^T \hat{\mathbf{X}}$, then

$$D_{\hat{\mathbf{x}}} = T^2 \quad (26)$$

where T^2 is defined in (24). Lemma 2 is proven in Appendix B. With this lemma, we can compare $\hat{\mathbf{x}}$ and $\hat{\mathbf{x}}_s$ in the variable space instead of comparing \mathbf{t} and \mathbf{t}_s . According to (16) and (20), $\hat{\mathbf{x}}$ and $\hat{\mathbf{x}}_s$ are projections of \mathbf{x} onto the same subspace, $\text{Span}\{\mathbf{P}\}$, along $\text{Span}\{\mathbf{R}\}^\perp$ and $\text{Span}\{\mathbf{P}\}^\perp$, respectively. Score \mathbf{t} reflects the variations correlated to \mathbf{Y} , while \mathbf{t}_s reflects only parts of variations correlated to \mathbf{Y} .

5.2.2. Relation between $\tilde{\mathbf{x}}$ and $\tilde{\mathbf{x}}_s$

From (16) and (20) we know that $\tilde{\mathbf{x}}_s$ and $\tilde{\mathbf{x}}$ are projections of \mathbf{x} onto $\text{Span}\{\mathbf{P}\}^\perp$ and $\text{Span}\{\mathbf{R}\}^\perp$, respectively, along the same subspace $\text{Span}\{\mathbf{P}\}$. Thus,

$$\tilde{\mathbf{x}}_s = \Pi_P^\perp \tilde{\mathbf{x}} \quad (27)$$

where Π_P^\perp is the orthogonal projection onto $\text{Span}\{\mathbf{P}\}^\perp$. If we use $Q = \|\tilde{\mathbf{x}}\|^2 \leq \delta_\alpha^2$ as in PLS based monitoring (δ_α^2 is the control limit of Q), the control region is a hyper-sphere in $\text{Span}\{\mathbf{P}\}^\perp$. When projected onto $\text{Span}\{\mathbf{P}\}^\perp$, it becomes a hyper-ellipsoid in the $\text{Span}\{\mathbf{P}\}^\perp$. If we use $Q_s = \|\tilde{\mathbf{x}}_s\|^2 \leq \delta_{s,\alpha}^2$ ($\delta_{s,\alpha}^2$ is the control limit of Q_s) in SIMPLS based monitoring, the control region is a hyper-sphere in the $\text{Span}\{\mathbf{P}\}^\perp$. The above differences can affect the monitoring results.

5.3. Monitoring with W-PLS

W-PLS calculates the scores and residuals as follows:

$$\begin{aligned} \mathbf{t}_{w,new} &= \mathbf{W}^T \mathbf{x}_{new} \\ \tilde{\mathbf{x}}_{w,new} &= (\mathbf{I} - \mathbf{W}\mathbf{W}^T) \mathbf{x}_{new}. \end{aligned} \quad (28)$$

Statistics T_w^2 and Q_w are defined similar to (24).

5.3.1. Relation between \mathbf{t} and \mathbf{t}_w

From (4),

$$\mathbf{W} = \mathbf{R}\mathbf{G} \quad (29)$$

where $\mathbf{G} = \mathbf{P}^T \mathbf{W}$ is nonsingular. Substituting (29) into (28) and (24), and considering (23), we have

$$\mathbf{A}_w = \frac{1}{n-1} \mathbf{T}_w^T \mathbf{T}_w = \mathbf{G}^T \frac{1}{n-1} \mathbf{T}^T \mathbf{T} \mathbf{G} = \mathbf{G}^T \mathbf{A} \mathbf{G} \quad (30)$$

and

$$T_w^2 = \mathbf{t}_w^T \mathbf{A}_w^{-1} \mathbf{t}_w = \mathbf{t}^T \mathbf{G} \mathbf{A}_w^{-1} \mathbf{G}^T \mathbf{t} = \mathbf{t}^T \mathbf{A}^{-1} \mathbf{t} = T^2. \quad (31)$$

Therefore, the monitoring results using \mathbf{t} and \mathbf{t}_w are the same.

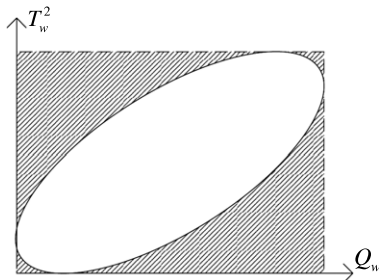


Fig. 3. Monitoring with T_w^2 , Q_w together.

Table 2
Properties of different monitoring policies.

Monitoring policy	$\mathbf{T} \perp \tilde{\mathbf{X}}$	$\hat{\mathbf{x}} \perp \tilde{\mathbf{x}}$	\mathbf{t} is \mathbf{Y} -related
PLS	Yes	No	Yes
W-PLS	No	Yes	Yes
SIMPLS	No	Yes	No

5.3.2. Relation between $\tilde{\mathbf{x}}$ and $\tilde{\mathbf{x}}_w$

From (16) and (19), $\tilde{\mathbf{x}}$ and $\tilde{\mathbf{x}}_w$ are projections of \mathbf{x} onto $\text{Span}\{\mathbf{R}\}^\perp$ along $\text{Span}\{\mathbf{P}\}$ and $\text{Span}\{\mathbf{R}\}$, respectively. Therefore, Q and Q_w monitor variabilities projected along different subspaces. The effect of these projections are studied later in this paper via simulation.

5.4. Comparison of three PLS monitoring policies

Table 2 shows properties of the PLS monitoring policies. The first and second properties are explained in (21) and (22). The third property describes whether T^2 explains \mathbf{Y} -related variations.

For monitoring using W-PLS, there are two shortcomings as $\tilde{\mathbf{x}}_w$ is correlated to \mathbf{t}_w . First, Q_w will be affected by abnormal variations in S_p . Thus, it may cause reductant alarms in Q_w , even though the fault is in S_p only, which makes it difficult to perform further fault diagnosis. Secondly, if we monitor process with Q_w and T_w^2 together, the chance of failing to alarm will increase, which is shown in Fig. 3. Since T_w^2 and Q_w are correlated, the normal condition should be located inside the ellipse, but the normal region defined by T_w^2 and Q_w is rectangle. Therefore, if a faulty sample is located in the shaded area in Fig. 3, the W-PLS policy fails to alarm.

Similarly, monitoring using SIMPLS suffers from the same drawbacks. In addition, since \mathbf{t}_s is not completely \mathbf{Y} -related, SIMPLS based monitoring loses the original purpose of PLS monitoring.

For monitoring using PLS, although $\hat{\mathbf{x}}$ is not orthogonal to $\tilde{\mathbf{x}}$, we can replace $\hat{\mathbf{x}}$ with $\hat{\mathbf{x}}_w$ according to Lemma 2 and (31) without any change of the monitoring results. From (16) and (20), it can be seen that $\hat{\mathbf{x}}_w \perp \tilde{\mathbf{x}}$. Thus, the standard PLS is the most appropriate for process monitoring among various algorithms.

6. Case study on simulation examples

6.1. A numerical example

Consider the following numerical example first:

$$\begin{cases} \mathbf{x}_k = \mathbf{A}\mathbf{z}_k + \mathbf{e}_k \\ \mathbf{y}_k = \mathbf{C}\mathbf{x}_k + v_k \end{cases} \quad (32)$$

where $\mathbf{A} = \begin{pmatrix} 1 & 4 & 4 \\ 2 & 0 & 1 \end{pmatrix}^\top$, $\mathbf{z}_k \sim \mathbf{N}(\mathbf{0}, 0.5^2\mathbf{I}_2)$, $\mathbf{e}_k \sim \mathbf{N}(0, 0.05^2\mathbf{I}_3)$, $v_k \sim \mathbf{N}(0, 0.05^2)$, $\mathbf{C} = \begin{pmatrix} 2 & 2 & 1 \end{pmatrix}$. The fault is added in the following form:

$$\mathbf{x}_k = \mathbf{x}_k^* + \boldsymbol{\Xi}f \quad (33)$$

Table 3
Fault occurs only in S_p .

f	Fault detection rates (%)					
	T^2	T_w^2	T_s^2	Q	Q_w	Q_s
0	1.1	1.1	1.1	1.4	1.6	1.4
5	25.0	25.0	25.0	1.5	2.1	1.5
std	0.6	0.6	0.6	0.18	0.18	0.18
10	85.0	85.0	85.0	1.7	3.4	1.7
std	0.5	0.5	0.5	0.16	0.22	0.16
15	99.5	99.5	99.5	1.4	4.4	1.4
std	0.07	0.07	0.07	0.18	0.29	0.18
20	100	100	100	1.6	6.6	1.6
std	0.003	0.003	0.003	0.15	0.37	0.15
25	100	100	100	1.6	10.3	1.6
std	0	0	0	0.16	0.48	0.16

Table 4
Fault occurs only in S_r .

f	Fault detection rates (%)					
	T^2	T_w^2	T_s^2	Q	Q_w	Q_s
0	1.2	1.2	1.2	1.5	1.5	1.5
2	1.3	1.3	1.3	32.5	31.4	32.5
std	0.17	0.17	0.16	0.76	0.71	0.76
4	1.0	1.0	0.9	96.4	95.9	96.4
std	0.17	0.17	0.17	0.27	0.27	0.27
6	1.0	1.0	1.0	100	100	100
std	0.17	0.17	0.16	0	0	0
8	1.0	1.0	1.0	100	100	100
std	0.17	0.17	0.16	0	0	0
100	1.2	1.2	7.9	100	100	100
std	0.15	0.15	0.40	0	0	0

where \mathbf{x}^* is the normal value produced by (32), $\boldsymbol{\Xi}$ is the fault direction vector, and f is the fault magnitude.

We use 100 samples under normal operation condition to perform a PLS model with one component. Given a fault direction and magnitude, 50 faulty samples per batch are produced according to (33) for fault detection. One hundred Monte Carlo simulations are performed to obtain the fault detection rates for PLS, WPLS and SIMPLS. The confidence of control limits is set to 99%. In order to show the efficiency of the comparison, we repeat the above procedure for 100 times and calculate the standard deviations of the detection rates, denoted by 'std' in the results.

6.1.1. Fault occurs in S_p only

Let $\boldsymbol{\Xi}_1 = [0.2500, 0.6641, 0.7046]^\top \in \text{Span}\{\mathbf{P}\}$, where $\|\boldsymbol{\Xi}_1\| = 1$. Thus, the fault occurs in S_p only. Table 3 shows the detection rates with fault magnitudes from 0 to 25.

From Table 3, it is observed that as the fault magnitude grows, the detection rates by T^2 , T_s^2 , T_w^2 increase, up to 100%. Q and Q_s are not affected by this fault, while Q_w alarms redundantly. Q_s has the same result as Q , which indicates the overlapped part monitored by T_s^2 and Q_s is not affected by this fault. 'std' is the standard deviation of the estimated fault detection rate. From Table 3, it is observed that the differences among detection rates are significant compared to the corresponding standard deviations.

6.1.2. Fault occurs in S_r only

Let $\boldsymbol{\Xi}_2 = [0.9582, -0.1931, -0.2111]^\top \in \text{Span}\{\mathbf{R}\}^\perp$, where $\|\boldsymbol{\Xi}_2\| = 1$. Thus, the fault occurs in S_r only. Table 4 shows the fault detection rates of this fault with fault magnitudes from 0 to 8 and 100.

From Table 4, we observe that T^2 and T_w^2 are not affected by this fault. When fault magnitude is very large, T_s^2 is affected. This is because \mathbf{t}_s is correlated to $\tilde{\mathbf{x}}_s$. However, T_w^2 is identical to T^2 in the monitoring as analyzed above, which indicates that the overlapped part monitored by T_w^2 and Q_w is normal under this fault. It is also

Table 5
TEP case study.

IDV	Fault detection rates (%)								
	T^2	T_w^2	T_s^2	Q	Q_w	Q_s	$T^2 + Q$	$T_w^2 + Q_w$	$T_s^2 + Q_s$
0	3.0	3.0	1.9	2.9	2.6	2.5	5.6	5.5	4.4
2	97.6	97.6	96.6	98.3	98.3	98.3	98.5	98.5	98.3
6	99.5	99.5	99.3	100	100	100	100	100	100
8	95.5	95.5	93.0	96.8	96.8	96.8	97.6	97.6	97.5
12	97.6	97.6	96.7	98.3	97.8	98.3	99.1	99.1	98.8
3	2.3	2.3	1.8	2.8	1.5	2.3	4.6	3.8	3.8
9	5.4	5.4	3.9	6.4	5.5	6.3	11.3	10.5	9.8
11	29.0	29.0	16.6	64.6	61.1	64.5	68.1	64.8	66.3
15	4.1	4.1	3.3	2.0	1.2	1.7	6.1	5.3	5.1

observed that as the fault magnitude grows, the detection rates by Q , Q_s , Q_w increase, and Q_w has a lower detection rate than Q and Q_s . The difference among detection rates is significant considering their standard deviations from Table 4.

6.1.3. Summary on the numerical example

The comparison of detection rates for faults in both S_y and S_r is not included due to page limitation, which has a similar result to the case study in the next subsection.

For all nonzero fault magnitudes, fault detection rates also reflect missing alarm rates. Further, the detection rates for zero fault magnitude correspond to the false alarm rates. From Tables 3 and 4, it can be observed that the false alarm rates for three policies are nearly the same.

6.2. Case study on TEP

In order to illustrate our point of view further, we use the Tennessee Eastman Process (TEP) as a case study (Downs & Vogel, 1993). The process is operated under closed-loop control. TEP has been widely used as a benchmark process for evaluating the process diagnosis methods such as PCA, multi-way PCA, and Fisher discriminant analysis (FDA) (Chiang, Russell & Braatz, 2001). PLS-based method has also been applied to the TEP (Lee et al., 2004). The TEP contains two blocks of variables: 12 manipulated variables and 41 measured variables (Chiang et al., 2001; Lee et al., 2004). Manipulated variables and process measurements are sampled with interval of 3 min. 19 composition measurements are sampled with time delays that vary from 6 min to 15 min. This time delay has a potentially critical impact on product quality control within the plant, which implies that the fault effect on product quality cannot be detected until the next sample of \mathbf{Y} is available.

In this study, the composition of G in Stream 9 is chosen as the quality variable \mathbf{y} with a time delay of 6 min. 22 process measurements and 11 manipulated variables are chosen as \mathbf{X} . There are 15 types of known faults in TEP, which are represented as IDV1–15. The detailed definition of \mathbf{X} , \mathbf{y} variables and all the faults can be found in Chiang et al. (2001). First, 480 normal samples are centered to zero mean and scaled to unit variance to build a PLS with $A = 6$. Then, 800 faulty samples for each fault are used for fault detection based on three policies. The fault detection rates for several typical faults are listed in Table 5.

In Table 5, IDV 0 is normal data and the corresponding detection rate is the false alarm rate. IDV 2,6,8,12 are the faults related to quality data \mathbf{y} , and IDV 3,9,11,15 are the faults unrelated to quality data \mathbf{y} (Zhou, Li & Qin, 2009). '+' means logical connection 'or'. From Table 5, we can find that PLS is more sensitive to detect the faults related to quality data \mathbf{y} , which is the purpose of monitoring with PLS. It is also observed that standard PLS has higher detection rates than W-PLS and SIMPLS in these simulated cases.

7. Conclusions

PLS has been widely used for monitoring complex industrial processes when quality variables are taken into account. There is, however, a lack of understanding of PLS geometry for the purpose of process monitoring. In this paper, the effect of \mathbf{Y} on the decomposition of the \mathbf{X} -space is clearly shown and the geometric interpretation of the PLS decomposition structure is given. Based on this interpretation, two alternative algorithms of PLS, W-PLS and SIM-PLS, are compared with the standard PLS in terms of the latent space decomposition and process monitoring. It is demonstrated that orthogonal sample space decomposition achieved by PLS is critical for process monitoring. It is concluded from analysis and simulation that monitoring using W-PLS and SIMPLS will cause ambiguous alarms and more missed alarms than the standard PLS. The standard PLS is the most appropriate for process monitoring among these alternative algorithms.

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Appendix A. Proof of Lemma 1

According to Zhang (2004), the oblique projector onto $\text{Span}(\mathbf{H})$ along $\text{Span}(\mathbf{S})$ can be obtained by the following equation generally:

$$\Pi_{H|S} = \mathbf{H}(\mathbf{H}^T \Pi_S^\perp \mathbf{H})^{-1} \mathbf{H}^T \Pi_S^\perp \quad (\text{A.1})$$

where Π_S^\perp is the orthogonal projector onto the $\text{Span}(\mathbf{S})^\perp$. Noting that

$$\Pi_{R^\perp}^\perp = \Pi_R = \mathbf{R}(\mathbf{R}^T \mathbf{R})^{-1} \mathbf{R}^T \quad (\text{A.2})$$

Substituting (A.2) into (A.1) and considering (5), \mathbf{R} is full-column-ranked, we can obtain

$$\Pi_{P|R^\perp} = \mathbf{P} \mathbf{R}^T \quad (\text{A.3})$$

Similarly, we have

$$\Pi_{R^\perp|P} = \mathbf{I} - \mathbf{P} \mathbf{R}^T \quad (\text{A.4})$$

Lemma 1 is proven. \square

Appendix B. Proof of Lemma 2

For a new sample \mathbf{x}_{new} , expanding T^2 and $D_{\hat{x}}$, we have

$$T^2 = \mathbf{x}_{new}^T \mathbf{R} \mathbf{A}^{-1} \mathbf{R}^T \mathbf{x}_{new} \quad (\text{B.1})$$

$$D_{\hat{x}} = \mathbf{x}_{new}^T \mathbf{R} \mathbf{P}^T (\mathbf{P} \mathbf{A} \mathbf{P}^T)^{-1} \mathbf{P} \mathbf{R}^T \mathbf{x}_{new} \quad (\text{B.2})$$

As \mathbf{P} is a full-column-ranked matrix, there exist two orthonormal matrix $\mathbf{U} \in \mathbb{R}^{m \times m}$, $\mathbf{V} \in \mathbb{R}^{A \times A}$, so that

$$\mathbf{P}^T = \mathbf{V} [\boldsymbol{\Sigma}, \mathbf{O}] \mathbf{U}^T \quad (\text{B.3})$$

where $\boldsymbol{\Sigma}$ is a nonsingular diagonal matrix, and \mathbf{O} is a zero matrix. Then, we have

$$\begin{aligned} \mathbf{P}^T (\mathbf{P} \mathbf{A} \mathbf{P}^T)^{-1} \mathbf{P} &= \mathbf{V} [\boldsymbol{\Sigma}, \mathbf{O}] \mathbf{U}^T (\mathbf{P} \mathbf{A} \mathbf{P}^T)^{-1} \mathbf{U} [\boldsymbol{\Sigma}, \mathbf{O}]^T \mathbf{V}^T \\ &= \mathbf{V} [\boldsymbol{\Sigma}, \mathbf{O}] ([\boldsymbol{\Sigma}, \mathbf{O}]^T \mathbf{V}^T \mathbf{A} \mathbf{V} [\boldsymbol{\Sigma}, \mathbf{O}])^{-1} [\boldsymbol{\Sigma}, \mathbf{O}]^T \mathbf{V}^T \\ &= \mathbf{V} [\boldsymbol{\Sigma}, \mathbf{O}] \begin{bmatrix} (\boldsymbol{\Sigma} \mathbf{V}^T \mathbf{A} \mathbf{V} \boldsymbol{\Sigma})^{-1} & \mathbf{O} \\ \mathbf{O} & \mathbf{O} \end{bmatrix} [\boldsymbol{\Sigma}, \mathbf{O}]^T \mathbf{V}^T \\ &= \mathbf{V} \boldsymbol{\Sigma} (\boldsymbol{\Sigma} \mathbf{V}^T \mathbf{A} \mathbf{V} \boldsymbol{\Sigma})^{-1} \boldsymbol{\Sigma} \mathbf{V}^T = \mathbf{A}^{-1}. \end{aligned} \quad (\text{B.4})$$

Lemma 2 is proven. \square

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