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用無母數迴歸方法監控曲線型品質特性之製程



研究生:翁崢珮

指導教授:洪志真 教授

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Profile Monitoring by Nonparametric Regression

研究生:翁崢珮 Student: Zheng-Pei Weng

指導教授:洪志真教授 Advisor: Dr. Jyh-Jen Horng Shiau

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摘要

此篇論文的重點在,對品質特性為一個函數或一條曲線的製程, 我們提出一些新的控制圖來做監控;除了平滑的假設外,在此不限定 曲線的形式。

我們提出了三個新的監控方法:(1)用無母數迴歸方法 "B-spline"對曲線做配適,再利用T²控制圖對製程做監控;(2)用典 型的控制圖EWMA、EWMSD或R控制圖對樣本曲線和參考曲線間的殘差做 監控;(3)用事先制定的測度(metrics)及其臨界值對品質特性曲線做 監控。

經過多次的模擬,利用平均連串長度(ARL)來比較所有方法,結 果顯示在曲線型品質特性為指數函數時,我們所提出的新方法都表現 得相當良好。最後在面對製程的各種偏移情形,該使用何種控制圖做 監控,我們也提出了一些建議。

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Student : Zheng-Pei Weng

Advisor : Dr. Jyh-Jen Horng Shiau

Institute of Statistics National Chiao Tung University

ABSTRACT

In this paper, we propose and study some control chart methods for monitoring processes for which the quality is characterized by a profile or a function. No assumptions are made on the form of the function except that it represents a smooth curve.

Three approaches of monitoring schemes are proposed: (1) Use the "B-spline" smoothing method to fit the profiles of processes and design a T^2 chart to monitor the deviations of the sample profiles from the in-control reference profile. (2) Use typical control charts – the EWMA, or EWMSD, or R chart to monitor the residuals between the sample profiles and the in-control reference profile. (3) Use some metrics defined to measure the deviations of the sample profiles from the in-control reference profile to monitor the sample profiles.

We construct a simulation study using an exponential profile as an example to investigate the effectiveness of the proposed schemes. The performances of these schemes are evaluated and compared in terms of the average run length. The results of the simulation study show that all approaches appear to perform well for exponential profiles. It is found that some control schemes are preferable to others for particular types of process changes.

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1. Introduction

Statistical process control (SPC) has been successfully applied in many domains, especially in industries. We often use control charts to monitor processes in most SPC applications. In the past, it was always assumed that the quality of a product or process can be measured by one or multiple quality characteristics. However, in many practical situations, the quality of a product or process is better characterized by a profile or a function. One example involved aspartame (an artificial sweetener), which is characterized by the amount that dissolves per liter of water at different levels of temperature (Kang and Albin, 2000). For illustration purpose, Figure 1 gives an example of 4 hypothetical sample profiles created based on the Figure 1 of Kang and Albin (2000). Another example is a semiconductor manufacturing application involving the calibration of a mass flow controller in which the performance of the process is characterized by a linear function (Mestek et al., 1994). Some profiles monitoring methods are proposed for this type of processes. Walker and Wright (2002) 411111 proposed additive models to assess the sources of variation of the density profile of particleboards. Jin and Shi (2001) used wavelets to monitor and diagnose process faults. Kang and Albin (2000) and Kim, Mahmoud, and Woodall (2003) proposed some methods for monitoring the process in which the performance is characterized by a linear profile.

Kang and Albin (2000) presented two approaches to monitor linear profiles. The first approach uses a multivariate T^2 control chart to monitor the profile parameters, slope and intercept, simultaneously. The second approach treats the residuals of the sample profile (from the reference profile) as a subgroup and uses a combined EWMA/R (exponentially weighted moving average/range) chart for profile monitoring. Kim et al. (2003) presented another approach. First, for each profile, code

the X-values so that the \overline{X} is zero and the estimators of the intercept and the slope of the regression line are independent. Then, construct two two-sided EWMA charts to monitor the intercept and the slope separately, and use a one-sided EWMA chart to monitor the process variation. This scheme is called EWMA₃. We remark that these two papers only addressed liner profiles.

The main objective of this paper is to propose and study profile monitoring schemes for profiles of more flexible shapes such that the schemes can be applied to more general and practical situations. To model a profile with no restriction on the functional form, it is natural to consider the nonparametric regression model in which the function is only assumed to be smooth. In this paper, for the profile modeling, we adopt the spline regression as the curve fitting/smoothing technique for the simplicity and readiness of a direct extension from the simple linear regression used by Kang and Albin (2000) and Kim et al. (2003) in constructing the T^2 chart. The second approach we consider is a monitoring scheme based on the residuals of sample profiles from the reference profile, similar to the combined EWMA/R scheme given in Kang and Albin (2000) and the combined EWMA₃ scheme by Kim et al. (2003). But instead of combining several control charts, we consider the EWMA chart for detecting the mean shift, the R chart for variation change, and the exponentially weighted moving standard deviation (EWMSD) chart for the variation increase separately. Since each of these control charts is designed for a particular type of process changes, their detecting powers usually are better than those more general-purposed charts. The 3rd approach is to use some metrics defined to measure the deviations between observed profiles and the reference profile to detect process changes. This method was firstly presented in Gardner et al. (1997), in which spatial signature metrics were used to diagnose the equipment faults. They reported that these metric-related charts are very powerful in detecting standard deviation shifts. Finally, a simulation study is conducted to compare the performances of all the proposed methods in terms of the average run length (ARL).

The remaining of this paper is organized as follows. Section 2 reviews the monitoring schemes proposed by Kang and Albin (2000) and Kim et al. (2003). Section 3 describes the three proposed approaches in details. Section 4 presents the results of a simulation study with some discussions and recommendations. Finally, Section 5 concludes the paper with a brief summary and discussion.

2. Literature Review

Let { $(x_i, y_{ij}), i = 1, 2, ..., n$ } be the observations of the j^{th} profile, j = 1, 2, ...Kang and Albin (2000) and Kim et al. (2003) assumed that, when the process is in statistical control, the underlying model is

$$y_{ij} = A_0 + A_1 x_i + \varepsilon_{ij}, \quad i = 1, 2, ..., n$$
, (1)

where the ε_{ij} 's are independent and identically distributed (i.i.d.) normal random variables with mean zero and variance $\sigma^2 > 0$.

In establishing a monitoring scheme, there are usually two phases: Phase I and Phase II. The interests in Phase I are on understanding the process variation and estimating the in-control process parameters from historical data in order to set up appropriate control limits. In Phase II, we then use the control limits established from Phase I to monitor the on-line process data successively. The multivariate T^2 and the residual control charts designed for Phase II by the two sets of authors mentioned above are described below.

Kang and Albin (2000) proposed two control strategies to monitor the process. The first strategy is to use the bivariate T^2 chart to examine the regression coefficients. Assume that the in-control values of the parameters A_0 , A_1 , and σ^2 in Equation (1) are known. The least-squares estimators of A_0 and A_1 for the j^{th} sample profile are, respectively,

$$a_{0j} = \overline{y}_j - a_{1j}\overline{x} \text{ and } a_{1j} = \frac{S_{xy(j)}}{S_{xx}}, \qquad (2)$$

where $\overline{y}_j = n^{-1} \sum_{i=1}^n y_{ij}, \ \overline{x} = n^{-1} \sum_{i=1}^n x_i, \ S_{xy(j)} = \sum_{i=1}^n y_{ij}(x_i - \overline{x}), \text{ and}$
$$S_{xx} = \sum_{i=1}^n (x_i - \overline{x})^2.$$
 The estimators a_{0j} and a_{1j} jointly follow the bivariate normal

distribution with the mean vector

$$\boldsymbol{\mu} = (A_0, A_1)' \tag{3}$$

and the variance-covariance matrix



where

and

$$\sigma_{01} = -\sigma^2 \frac{\overline{x}}{S_{xx}}$$

It can be seen that a_{0j} and a_{1j} are dependent. We express the sample slope and the sample intercept in Equation (2) as the vector $\mathbf{z}_j = (a_{0j}, a_{1j})'$. Then, the T² statistic for the *j*th sample is given by

$$T_j^2 = (\mathbf{z}_j - \boldsymbol{\mu})' \boldsymbol{\Sigma}^{-1} (\mathbf{z}_j - \boldsymbol{\mu}), \qquad (5)$$

which follows the chi-square distribution with 2 degrees of freedom. Thus, the upper control limit is UCL = $\chi^2_{2,\alpha}$, where $\chi^2_{2,\alpha}$ is the 100(1- α) percentile of the chi-square distribution with 2 degrees of freedom. When there are shifts from the nominal values,

 A_0 and A_1 , Kang and Albin (2000) pointed out that T_j^2 in Equation (5) follows the non-central chi-square distribution with the parameter of noncentrality

$$\tau = (\lambda + \beta \overline{x})^2 n + \beta^2 S_{xx}$$
(6)

when the intercept A_0 shifts to $A_0 + \lambda \sigma$ and the slope A_1 shifts to $A_1 + \beta \sigma$.

One disadvantage of the T^2 chart is that it cannot distinguish the sources of the process change --- mean or variance. Kang and Albin (2000) then proposed another approach to monitor the mean and variance separately with two charts. The second strategy is to use an EWMA chart to monitor the average of the regression residuals and simultaneously an R chart to monitor the range of the residuals. One reason for adding the R chart to the EWMA chart is that the EWMA chart is not sensitive to shifts in the process variation. Another reason is that the EWMA chart based on the residual average is not sensitive to some shifts in A_0 and A_1 , for which the magnitudes of the residuals tend to be large, but the residual average tends to be very small. Details of these two charts are reviewed below.

The regression residuals for the j^{th} sample are

$$e_{ij} = y_{ij} - A_0 - A_1 x_i$$
, $i = 1, 2, ..., n$. (7)

The average of the residuals for sample *j* is

$$\overline{e}_j = \frac{\sum_{i=1}^n e_{ij}}{n}.$$

Then, the EWMA statistic for sample *j* is given by

$$z_j = \theta \overline{e}_j + (1 - \theta) z_{j-1} ,$$

where θ (0 < $\theta \le 1$) is a smoothing constant and $z_0 = 0$. The control limits for the EWMA chart are

$$LCL = -L\sigma \sqrt{\frac{\theta}{(2-\theta)n}}$$
 and $UCL = L\sigma \sqrt{\frac{\theta}{(2-\theta)n}}$, (8)

where *L* is a constant selected for the control chart to achieve a prescribed in-control ARL. If z_j is smaller than LCL or greater than UCL, it is regarded as an out-of-control signal. For the R chart, the R statistic for the *j*th sample is given by

$$R_i = \max_i(e_{ii}) - \min_i(e_{ii}).$$

The control limits are

$$LCL = \sigma(d_2 - L_R d_3) \text{ and } UCL = \sigma(d_2 + L_R d_3), \qquad (9)$$

where L_R is also a positive constant chosen to achieve a specified false alarm rate, d_2 and d_3 are constants relating the range and the standard deviation, which depend on the subgroup size *n*. Commonly used values of d_2 and d_3 can be found in Montgomery (2001). The process is claimed out of control when any of the EWMA and the R statistics is out of control.

Kang and Albin (2000) used simulations to demonstrate the effectiveness of these two approaches. They reported that the out-of-control ARL of the combined EWMA/R chart for the residuals is smaller than that of the T^2 chart for shifts in the intercept and the slope. On the other hand, the multivariate T^2 chart requires only one chart.

Kim et al. (2003) first coded the X-values by centering so that the average of the coded X-values is zero. Then, the least-squares estimators of the slope and intercept are independent. See Myers (1990) and Ryan (1997). Accordingly, transform the model in Equation (1) into

$$y_{ii} = B_0 + B_1 x_i^* + \varepsilon_{ii}, \quad i = 1, 2, ..., n,$$
 (10)

where $B_0 = A_0 + A_1 \overline{x}$, $B_1 = A_1$, and $x_i^* = x_i - \overline{x}$. For the j^{th} sample, the least-squares

estimator of B_0 is $b_{0j} = \overline{y}_j$. The estimator of B_1 is the same as that of A_1 . Because b_{0j} and b_{1j} are independent, they are normally distributed with means B_0 , B_1 and variances σ^2 / n , σ^2 / S_{xx} , respectively. Consequently, the authors recommended three EWMA charts to monitor the slope, the intercept, and the error variance, respectively. The EWMA statistic for the intercept parameter B_0 is

$$EWMA_{I}(j) = \theta b_{0j} + (1 - \theta) EWMA_{I}(j - 1)$$
⁽¹¹⁾

with θ (0 < $\theta \le 1$) being a smoothing constant and *EWMA*₁(0) = B₀. The control limits are

$$LCL = B_0 - L_I \sigma \sqrt{\frac{\theta}{(2-\theta)n}}$$
 and $UCL = B_0 + L_I \sigma \sqrt{\frac{\theta}{(2-\theta)n}}$, (12)

where L_i is a constant chosen for achieving a prescribed in-control ARL. When $EWMA_i(j) < LCL$ or $EWMA_i(j) > UCL$, an out-of-control signal is alarmed. The EWMA statistic for the slope parameter B_i is

$$EWMA_{s}(j) = \theta b_{1j} + (1 - \theta) EWMA_{s}(j - 1)$$
(13)

with θ (0 < $\theta \le 1$) being a smoothing constant and *EWMA*_S(0) = *B*₁. The lower and upper control limits are, respectively,

$$LCL = B_1 - L_S \sigma \sqrt{\frac{\theta}{(2-\theta)S_{xx}}}$$
 and $UCL = B_1 + L_S \sigma \sqrt{\frac{\theta}{(2-\theta)S_{xx}}}$, (14)

where L_s is a constant chosen in order to obtain a specified in-control ARL. They also used a one-sided EWMA chart to detect only increases in process variability. The EWMA statistic for that is given by

$$EWMA_{E}(j) = \max\{\theta \ln(MSE_{j}) + (1-\theta)EWMA_{E}(j-1), \ln(\sigma_{0}^{2})\}, \qquad (15)$$

where θ ($0 < \theta \le 1$) again is a smoothing constant, $EWMA_E(0) = \ln(\sigma_0^2)$, and

 $MSE_{j} = (n-2)^{-1} \sum_{i=1}^{n} (y_{ij} - b_{0j} - b_{1j} x_{i}^{*})^{2}$. They used the following approximation that is very similar to the result derived by Crowder and Hamilton (1992) to construct the control limit:

$$Var[ln(MSE_j)] \approx \frac{2}{n-2} + \frac{2}{(n-2)^2} + \frac{4}{3(n-2)^3} - \frac{16}{15(n-2)^5}.$$

Then an upper control limit is given by

$$UCL = L_E \sqrt{\frac{\theta}{(2-\theta)} Var[\ln(MSE_j)]} , \qquad (16)$$

where L_E is again a constant selected to give a specified in-control ARL. The process is claimed out of control if any of the three statistics given in (11), (13), and (15), respectively is beyond the control limits.

Kim et al. (2003) showed by a simulation study that their methods are generally more effective and seem much more interpretable than the methods of Kang and Albin (2000) in Phase II for detecting sustained shifts in either the intercept or the slope or increases in the error variance.

In practice, the in-control values of parameters are usually unknown. Common practice is to use some historical data to estimate these parameters and construct preliminary control charts during Phase I. Kang and Albin (2000) estimated the regression parameters of Equation (1) based on the historical data containing *k* sample profiles. For each profile, obtain estimates a_{0j} and a_{1j} of A_0 and A_1 , respectively.

Then the reference line can be estimated by $\hat{Y} = a_0 + a_1 X$ and $\hat{\sigma}^2 = MSE$, where

$$a_0 = \frac{\sum_{j=1}^k a_{0j}}{k}, \ a_1 = \frac{\sum_{j=1}^k a_{1j}}{k}, \text{ and } MSE = \frac{\sum_{j=1}^k MSE_j}{k}$$

with $MSE_j = (n-2)^{-1} \sum_{i=1}^{n} (y_{ij} - a_{0j} - a_{1j}x_i)^2$. The T² statistic is modified by

$$T_{0j}^{2} = \frac{k}{k-1} (\mathbf{z}_{j} - \mathbf{z})' \mathbf{S}^{-1} (\mathbf{z}_{j} - \mathbf{z}), \qquad (17)$$

where $\mathbf{z} = (a_0, a_1)'$ and **S** is an unbiased estimate of Σ with components

$$S_{11} = MSE(\frac{1}{n} + \frac{\overline{x}^2}{S_{xx}}), \quad S_{22} = \frac{MSE}{S_{xx}}, \text{ and } S_{12} = -MSE(\frac{\overline{x}}{S_{xx}}).$$

Kang and Albin (2000) proved that the $T^2/2$ statistic in Equation (17) has the F distribution. Thus, the upper control limit of the T^2 chart in Phase I is $UCL = 2F_{2,(n-2)k,\alpha}$. For EWMA and R charts in Phase I, the control limits are modified by substituting σ with \sqrt{MSE} . There are other methods to examine the historical data in Phase I. Stover and Brill (1998) proposed a Hotelling's T^2 approach that is similar to the T^2 method of Kang and Albin (2000). The distinction between the two is the estimate of the variance-covariance matrix. Another approach is to use a univariate chart based on the first principal component corresponding to the vectors containing the estimators of the intercept and slope. But Kim et al. (2003) advised against this principal component method since it is unable to detect the shifts in the direction perpendicular to the first principal component.

3. Methodologies

3.1 B-splines

In order to extend linear profiles to any smooth functions, a smooth curve fitting technique is needed for profile smoothing from noisy data. A polynomial function is not very flexible for approximating curves with different degrees of smoothness at different locations. One way to overcome this drawback is to use locally a polynomial approximation of low degree. Another way is to allow the derivatives of the approximating function to have discontinuities at certain locations. This can be accomplished by fitting piecewise polynomials or splines. Frequently, a cubic spline, i.e., a piecewise polynomial with continuous first two derivatives, is used for such approximation. Consider the nonparametric regression model

$$y_i = m(x_i) + \varepsilon_i, \quad i = 1, ..., n,$$
 (18)

where m(x) is a regression curve and ε_i are i.i.d. normally distributed with zero mean and common variance $\sigma^2 > 0$. In this paper, we adopt the B-spline regression method for its popularity and simplicity. See de Boor (1978) for the definition of the B-spline basis. The points where the derivatives of the approximation function could have discontinuities are called knots. Let $t_1, ..., t_{b+k}$ be the knots, where *b* is the number of bases and each polynomial is of order *k*. Then each basis of order *k* is a *k*-2 continuously differentiable function. Define $B_{l,k}$ as a B-spline basis that is nonzero only on the interval $(t_l, t_{l+k}), l = 1, 2, ..., b$. A B-spline curve of order *k* can be constructed as

$$P(x_i) = \sum_{l=1}^{b} c_l B_{l,k}(x_i) \quad \forall x_i \in [u, v], \ i = 1, ..., n,$$

where c_l 's are the unknown B-spline coefficients and *n* is the number of set points in the interval [u, v]. So we can modify Equation (18) to

$$y_{i} = \sum_{l=1}^{b} c_{l} B_{l,k}(x_{i}) + \varepsilon_{i}, \quad i = 1, ..., n.$$
(19)

The B-spline basis $B_{l,k}$ of order k can be defined as

$$\begin{cases} if \ k = 1, \quad B_{l,1}(t) = \begin{cases} 1 \ for \ t_l \le t < t_{l+1}, \\ 0 \ for \ t < t_l \ and \ t \ge t_{l+1}; \end{cases} \\ if \ k \ge 2, \quad B_{l,k}(t) = \frac{t - t_l}{t_{l+k-1} - t_l} B_{l,k-1}(t) + \frac{t_{l+k} - t}{t_{l+k} - t_{l+1}} B_{l+1,k-1}(t). \end{cases}$$

The following S-PLUS function "spline.des" generates a matrix of B-spline bases evaluated at points provided by users.

spline.des(knots, x, ord, derives),

where "knots" is the vector of knots for the spline, "x" represents x-coordinates at which to evaluate the spline basis functions, "ord" is the order of the spline, and

"derives" is the order of the derivative to evaluate at each of the points. If this vector is given, it must have the same length as x. The default is a vector of zeros of the same length as x. We show an example of B-spline bases in Figure 2.

After constructing B-spline bases, we use these normalized B-spline functions $B_{l,k}$ as the regression basis function. For the given knots, the spline regression method finds the best spline approximation via the following least squares regression:

$$\min_{\mathbf{c}} \sum_{i=1}^{n} \{ y_i - \sum_{l=1}^{b} c_l B_{l,k}(x_i) \}^2 , \qquad (20)$$

where $\mathbf{c} = (c_1, ..., c_b)'$. So the least-squares estimator of \mathbf{c} is

$$\hat{\mathbf{c}} = (\mathbf{B'B})^{-1}\mathbf{B'y}$$

where $\mathbf{y} = (y_1, ..., y_n)'$, $\hat{\mathbf{c}} = (\hat{c}_1, ..., \hat{c}_b)'$, and **B** is the design matrix with the $(i, l)^{\text{th}}$ element $B_{l,k}(x_i)$, l = 1, ..., b, i = 1, ..., n. Then $\hat{\mathbf{c}}$ has a multivariate normal distribution with the mean vector **c** and the variance-covariance matrix $\boldsymbol{\Sigma} = \sigma^2 (\mathbf{B'B})^{-1}$.

In phase II, it is assumed that the reference profile is known. Denote it by f(x). We now establish a B-spline representation of the reference profile. First obtain n pairs of data, $\{(x_i, f(x_i)), i = 1, ..., n\}$. Let **c** be the least squares solution of (20) with y_i replaced by $f(x_i)$. We then treat this vector **c** as the true in-control value. For the j^{th} sample profile, compute the vector of sample estimators $\hat{\mathbf{c}}_j = (\hat{\mathbf{c}}_{1j}, ..., \hat{\mathbf{c}}_{bj})'$, where $\hat{c}_{1j}, ..., \hat{c}_{bj}$ are the estimated sample B-spline coefficients. Then the T^2 statistic of the j^{th} sample is given by $T_j^2 = (\hat{\mathbf{c}}_j - \mathbf{c})' \Sigma^{-1} (\hat{\mathbf{c}}_j - \mathbf{c})$. But the upper control limit in phase II needs to be modified as $UCL = \chi^2_{b,\alpha}$. And in Phase I, we also use historical data containing k sample profiles to construct the control chart. Then the reference curve can be estimated by $\hat{\mathbf{y}} = \mathbf{B}\tilde{\mathbf{c}}$ with $\tilde{\mathbf{c}} = (\tilde{c}_1, ..., \tilde{c}_b)'$ and $\hat{\sigma}^2 = MSE$ where

$$\tilde{c}_{1} = \frac{\sum_{j=1}^{k} \hat{c}_{1j}}{k}, \dots, \tilde{c}_{b} = \frac{\sum_{j=1}^{k} \hat{c}_{bj}}{k}, \text{ and } MSE = \frac{\sum_{j=1}^{k} MSE_{j}}{k}$$
(21)

with $MSE_j = \|\mathbf{y}_j - \mathbf{B}\hat{\mathbf{c}}_j\|^2 / (n-b)$ and $\|\cdot\|$ denotes the Euclidean norm. The T^2 statistic for the j^{th} sample profile is then modified by

$$T_{0j}^{2} = \frac{k}{k-1} (\hat{\mathbf{c}}_{j} - \tilde{\mathbf{c}}) \, \mathbf{S}^{-1} (\hat{\mathbf{c}}_{j} - \tilde{\mathbf{c}}) \,, \qquad (22)$$

where $\mathbf{S} = \hat{\sigma}^2 (\mathbf{B'B})^{-1}$ is an unbiased estimate of Σ . Thus, the upper control limit of the T² chart in Phase I is $UCL = bF_{b,(n-b)k,\alpha}$.

The choice of the number of bases, as in the role of the smoothing parameter in any nonparametric regression methods, is an important issue. The boundary effect is another issue. We will address these two issues in Section 4.

3.2 Residual EWMA and R Charts

The second approach we propose is to use the EWMA and the R chart to monitor **1896** the residual average and the range, respectively, instead of the combined EWMA/R chart proposed by Kang and Albin (2000). Kang and Albin (2000) found that the value L = 3.1151 yields an in-control ARL of approximately 802 for the EWMA chart and an in-control ARL of approximately 261 for the R-chart. Although ARL₀ of their combined procedure is close to 200, the detecting power of the EWMA chart is unequal to that of the R chart. In order to increase the detecting powers of both the EWMA and the R charts, we decide to monitor the process by using the EWMA chart to detect mean shifts and the R chart to detect standard deviation shifts of the residuals. The regression residual vector for the j^{th} sample is

$$\mathbf{e}_{i} = \mathbf{y}_{i} - \hat{\mathbf{y}}_{i} = \{\mathbf{I} - \mathbf{B}(\mathbf{B}'\mathbf{B})^{-1}\mathbf{B}'\}\mathbf{y}_{i}.$$

The details of the EWMA and the R charts are the same as that described in Section 2.

3.3 Metrics

The third approach is to use some metrics to monitor the process. Gardner et al. (1997) presented a new methodology based on some "metrics" for the equipment fault detection. The equipment fault detection was not only in that it incorporates the use of integrated spatial information in a virtual wafer surface that was fitted by thin-plate splines, but also in that it can be used to detect and classify equipment faults at the same time. Their main focus was to use the differences between the observed and the expected virtual wafer surfaces to construct metrics which can be used to detect and classify equipment faults at the same time. Their main focus was to use the differences between the observed and the expected virtual wafer surfaces to construct metrics which can be used to detect and classify equipment faults. Thus, the following general and specific metrics were designed:

$$M1 = \int_{R} (g - T)^{2} dR \quad \text{(Squared metric)};$$

$$M2 = \int_{R} |g - T| dR \quad \text{(Absolute-value metric)};$$

$$M3 = \int_{R} (g - T)^{2} dR \quad \text{if } |g - T| > 1.5 \overset{o}{A} \quad \text{(Spec-limit metric)};$$

$$= 0 \quad otherwise$$

$$M4 = \int_{R} (g - T)^{2} dR \quad \text{if } (g - T) > 0$$

$$= -\int_{R} |g - T| dR \quad \text{if } (g - T) \le 0$$

(Square-above-absolute-value-below metric);

$$M5 = \int_{R} (g-T)^{2} dR \quad if \ (g-T) \le 0$$
$$= -\int_{R} |g-T| dR \quad if \ (g-T) > 0$$

(Square-below-absolute-value-above metric);

where g is a newly fitted thin-plate spline surface, T is the target surface, and R denotes the wafer surface region. Metrics 1-3 are general metrics used to detect the presence of an equipment fault, and Metrics 4-5 are specific metrics used to detect specific fault patterns. Gardner et al. (1997) proposed an alternate Bayesian (simulation) approach that can be taken to determine the null distribution of the

metrics. According to a procedure given in Green and Silverman (1994), assuming a Gaussian prior distribution, the posterior distribution of the thin-plate spline surface **g** has the following multivariate normal distribution:

$$\mathbf{g} \sim \mathrm{MVN} \left[\hat{\mathbf{g}}, \hat{\sigma}^2 \mathbf{A}(\hat{\lambda}) \right],$$

where $\hat{\mathbf{g}}$ is the vector of fitted values, $\hat{\sigma}^2$ is calculated as the residual sums of squares about the fitted curve divided by an effective degrees of freedom, and $\mathbf{A}(\hat{\lambda})$ is the projection matrix which maps the vector of observed values to their predicted values. Simulate *M* independent sets of *n* observations from the posterior multivariate normal distribution described above. A spline surface is fitted to each set of observations, and the metrics are calculated. As a result, M independent values are obtained from the null distribution of each metric. The $(100 \times \alpha)^{\text{th}}$ percentile of the M simulated metric values is the critical value for determining if the new curve is far from the target curve. For each newly observed curve, metrics are calculated and compared to their corresponding critical values in order to determine whether or not a fault is detected. In the experimental data, they used the average predicted surface of two wafers as the target surface. 5,000 observations were simulated using parametric bootstrapping from MVN[$\hat{\mathbf{g}}, \hat{\sigma}^2 \mathbf{A}(\hat{\lambda})$] where $\hat{\mathbf{g}}$ is the average of the predicted values using the thin-plate spline fitting for these two wafers, and $\hat{\sigma}^2 \mathbf{A}(\hat{\lambda})$ is the average of the covariance matrices from the thin-plate spline fitting for these two wafers. Then these simulated observations were used to obtain 5,000 metric values and the corresponding empirical $\alpha = 0.01$ "critical value" for each metric.

With the same idea, we select two different and popular metrics to monitor the process in all types of shifts. They are Squared metric and Absolute-value metric. Two metrics are defined in the following:

$$M1 = \sum_{i=1}^{n} \{g(x_i) - T(x_i)\}^2 \quad \text{(Squared metric)};$$
$$M2 = \sum_{i=1}^{n} |g(x_i) - T(x_i)| \quad \text{(Absolute-value metric)};$$

where g is a newly fitted B-spline curve and T is the reference profile. To construct the control charts, we need to find the critical value of the null distribution of each metric.

Since the distributions of these metrics are difficult to obtain, we use historical data (*k* sample profiles) to compute the reference profile in Phase I. Fit a B-spline to each of the *k* sample profiles. Denote the average of *k* regression estimators for the *l*th B-spline coefficient by \hat{c}_i . So the estimated reference profile $\sum_{l=1}^{b} \hat{c}_l B_{l,k}(x)$ and $\hat{\sigma}^2 = MSE$ are available by Equation (21). Simulate *M* sets of *n* observations from the following model:

$$y_{i} = \sum_{l=1}^{b} \hat{c}_{l} B_{l,k}(x_{i}) + \varepsilon_{i}, \quad i = 1, ..., n,$$
(23)

where $\varepsilon_i \stackrel{i.i.d.}{\sim} N(0, \hat{\sigma}^2)$. For each metric, compute the metric value for each of the simulated profiles. Let the $(100 \times \alpha)^{\text{th}}$ percentile of these *M* values of the metric be the critical value. The process is claimed out-of-control when the metric of the newly observed profile is greater than the critical value.

3.4 EWMSD Chart

The last approach is using exponentially weighted moving standard deviation (EWMSD) to monitor the process. It is very sensitive in detecting shifts in process variation, particularly when shifts are relatively small. Define the sample standard deviation of the residuals for the j^{th} sample as

$$s_{j} = \sqrt{\frac{\left\|\mathbf{y}_{j} - \hat{\mathbf{y}}_{j}\right\|^{2}}{n-b}}, \ j = 1, 2, \dots$$

The EWMSD statistic is given by

$$v_i = \theta s_i + (1 - \theta) v_{i-1},$$

where θ (0 < $\theta \le 1$) is a smoothing constant and $v_0 = \sigma_0$. In Phase II, assume the in-control value of σ^2 is known. The control limits for the EWMSD chart are

$$LCL = \sigma - L\sigma \sqrt{\frac{\theta}{(2-\theta)n}}$$
 and $UCL = \sigma + L\sigma \sqrt{\frac{\theta}{(2-\theta)n}}$

where L is also a constant chosen to give a specified in-control ARL. When the EWMSD statistic is beyond the control limits, the process is claimed out-of-control. In Phase I, the control limits are modified by substituting σ with \sqrt{MSE} .

4. Simulation Studies

We assess the performances of these approaches in terms of ARL through simulation studies. For all approaches except the metric method, we assume the underlying reference profile is known. Denote the in-control ARL value by ARL₀. All charts are designed to have the same $ARL_0 = 200$, which corresponds to $\alpha = 0.005$. For each metric under study, we simulate 50,000 in-control profiles to approximate the distribution of the metric. The critical value is set to be the 99.5th percentile of these simulated metric values such that the ARL₀ = 200. The smoothing constant θ may affect the ARL performances of the EWMA and EWMSD charts. Details can be seen in Lucas and Saccucci (1990). In our study, the smoothing constant is set to 0.2. In Phase II process monitoring, for each control chart, run lengths are generated for the in-control and various out-of-control situations. The run length is the number of profiles generated when the first out-of-control signal occurs. Generate profiles from the underlying model until the first out-of-control signal is alarmed to obtain a run length. Repeat N times to obtain N run lengths and then estimate the ARL by averaging these N run lengths. Denote the estimator by $A\hat{R}L$. To obtain the standard

error of ARL (the ARL estimator), one simple way is to compute the sample standard deviation of the *N* simulated run lengths and then divided it by \sqrt{N} . Another estimator of the standard error is

$$\frac{A\hat{R}L(1-\frac{1}{A\hat{R}L})^{1/2}}{\sqrt{N}}$$

since the run length has a geometric distribution with mean 1/ p and standard deviation $(1 - p)^{1/2} / p$ with p = 1 / ARL. In our simulation, we generate a total of N = 100,000 run lengths to estimate the ARL value. If we construct the control charts such that $A\hat{R}L_0$ is about 200, then the standard error of this estimator is about $200(1 - \frac{1}{200})^{1/2} / \sqrt{100,000} \approx 0.631$. We adopt the first approach in this study.

In this paper, we extend the linear profile to a functional-form-free smooth function to adapt to more general cases. To choose an underlying reference profile for our study, we mimic the aspartame curves in Figure 1. As a result, we take an exponential function as the reference profile. In our simulation study, consider the 40000 exponential profile of the form $Y = I_0 + M_0 e^{-N_0(x-1)^2} + \varepsilon$, where $\varepsilon \sim N(0,1)$. The in-control reference profile is $1+15e^{-(x-1)^2}$, which is displayed in Fiugre 3. We choose x_i -values of 0, 0.08,..., 3.92 (n = 50) in our simulation. Four different types of shifts are considered in the simulation study: I shift, M shift, N shift, and error variance increase. The curve varies in different ways with different varying coefficients. Figures 4-6 illustrate the effect of the shifts. In these figures, the solid lines represent the true function $1+15e^{-(x-1)^2}$, while the dotted lines represent the shifted curves. Figure 4 gives the curve that is 2 units upward by an I shift. The dotted line in Figure 5 is the curve of $1+18e^{-(x-1)^2}$, that is, the y-value is magnified with the same multiple at each point because of an M shift. Figure 6 shows a case of N shift, and the resulting curve seems narrower than the reference curve. Its function is $1+15e^{-1.5(x-1)^2}$.

For T² chart, we use the sequence (-1.2, -0.8, -0.4, 0, 0.4, 0.8, 1.2, 1.6, 2, 2.4, 2.8, 3.2, 3.6, 4, 4.4, 4.8, 5.2) as the knots with order 4 so that the number of B-spline bases is 13. The knots sequence is equidistant, and the thirteen cubic B-splines are shown in Figure 7. Approximate the true exponential function by spline regression with these 13 B-spline bases. The fitted curve (dashed line) is shown in Figure 8. The circles are the values on the exponential function and the dashed line goes through these values smoothly. So the B-spline approximation is very close to the true function. The average squared error of the approximation is 0.0000099498 and it is computed by $\frac{1}{n} \sum_{i=1}^{n} (f(x_i) - \sum_{i=1}^{b} \hat{c}_i B_{i,k}(x_i)).$

It is well known that boundary effect is a potential problem in smoothing methods. We use a simple simulation study for T^2 chart to illustrate the boundary effect that we encountered in our study. First, we fix the number of set points at 20, and change the number of B-spline bases to 5, 9, and 13 and find that the ARL₀ of T^2 chart are 197.976, 193.265, and 187.814, respectively. That is, when n = 20, the ARL₀ decreases as the number of B-spline bases increases and the boundary effect is more and more obvious. Because the number of set points 20 is small, the B-spline coefficients for the boundary bases are not accurate enough. It leads to a larger T^2 statistic and a smaller ARL₀. So we increase the number of set points to 50. Compute the average of 50,000 replications of each B-spline coefficient. The simulation results are shown in Table 1. The first row is the 13 coefficients of the exponential reference curve fitted by B-splines and these values are treated as the true coefficients. The second row gives the differences between the average of 50,000 replications and the true coefficient for the first simulation. The third and fourth rows give the results of the second simulation and the third simulation, respectively. From Table 1, we can see the variability of each coefficient estimate. The two largest differences occur at the

13th and the first coefficient, respectively, in every simulation. In order to achieve ARL₀ = 200, we omit the coefficients 1 and 13 in constructing the T² statistic. Naturally, the control limit of the T² chart is adjusted to $\chi^2_{11,0.005}$.

Instead of using the EWMA and R charts simultaneously to monitor the process, we use the EWMA chart to detect the mean shifts, say, *I*, *M*, or *N* shifts, and use the R chart to detect the shift in the error variance. For the metric method, use Squared metric and Absolute-value metric to detect all types of shifts. All simulation results are presented below.

Table 2 and Figure 9 give the ARL values and curves for shifts in I_0 in unit of σ , respectively. In Table 2, the smallest ARL values are marked with the deep gray and the second smallest are marked with the light gray. The EWMA chart and two metrics perform better than the T² chart. The EWMA chart detects all shifts faster than others and the Absolute-value metric is the second best.

Table 3 and Figure 10 show the ARL values and curves for M shifts, respectively. The EWMA chart also performs much better than the T² chart and two metrics over the entire range of shifts considered. We can see that the performances of two metrics and T² chart are very similar with the Absolute-value metric better for smaller shifts and T² chart better for larger shifts.

Table 4 and Figure 11 show the ARL values and curves for N shifts, respectively. Again, the detecting power of the EWMA chart is the largest and the T^2 chart is the smaller than the other charts. Secondly, the results of two metrics are similar and the Absolute-value metric is faster for smaller shifts, say, 0.14σ or less, and slower for larger shifts.

Table 5 and Figure 12 give the ARLs for shifts in the process standard deviation. The R chart performs slightly better than the T^2 chart and two metrics over the entire range of shifts. And the Squared metric is also slightly faster than the Absolute-value metric over the whole range of shifts. But the EWMSD chart performs much better than the other charts, because it is designed specially for detecting the standard deviation shifts. This demonstrates that the EWMA and EWMSD charts are more sensitive for all shifts than others, especially for smaller shifts.

Table 6 gives the ARL performances when simultaneous M and N shifts are considered. The T^2 chart is uniformly slower than other charts over the entire range of shifts. The EWMA chart performs the best with any combination of M and N shifts, and Absolute-value metric is the second best. Tables 7 and 8 present the simulation results with I and M, I and N combinations, respectively. Two tables also show similar results.

To investigate the effect of the number of set points (n) and the number of B-spline bases (b), another simulation study is conducted. Table 9 and Figure 13 give the ARLs of the T^2 chart with different number of bases, 5, 9, and 13, for *I* shifts. The results show that the detecting power of the T^2 chart increases as parameter b decreases. Table 10 and Figure 14 give the ARLs of the T^2 chart with different n = 30, 40, and 50 for I shifts. We can see that the ARL decreases as n increases with various sizes in I shifts. In order to improve the power of the T^2 chart, it seems that a smaller b and larger n is suggested. However, there is another concern – the squared error between the fitted profile estimated by B-splines and the reference profile. Although the detecting power of T^2 chart with n = 50 and b = 5 is larger, the average squared error comes to 0.4829616. From Figure 15 we can see that, with b = 5, the B-spline does not approximate the reference profile well. Thus, b should not be too small.

Table 11 and Figure 16 show that the ARL decreases as n increases for the EWMA chart with various shifts in M shifts. Tables 12-13 and Figures 17-18 show that the detecting power increases as n increases for all metrics with I shifts or standard deviation shifts.

Back to linear profiles, we compare the ARL performances of our approaches to that of the methods proposed by Kang and Albin (2000) and Kim et al. (2003). We simulate again by using the same control limits of each chart. The linear profile model used by Kang and Albin (2000) is $y_{ij} = 3 + 2x_i + \varepsilon_{ij}$, where $\varepsilon_{ij} \stackrel{i.i.d.}{\sim} N(0,1)$ with fixed x_i -values of 2, 4, 6, and 8. Three different types of shifts, intercept shifts, slope shifts and standard deviation shifts, are considered in their papers. From our simulation study, we notice that our ARL values of EWMA₃ are 1 more than the values given in Kim et al. (2003). We guess the discrepancy may come from the way of counting the run lengths. They probably counted the run lengths as the number of the profiles before the out-of-control signal occurs, so that all values are one less than ours.

Table 14 and Figure 19 give the ARL values and curves for intercept shifts, respectively. The EWMA chart performs much better than other charts for smaller shifts. The Absolute-value metric detects the larger shifts, say, 1.4σ or more, faster than the other charts. We can clearly see the performance of EWMA₃ is better than that of the methods of Kang and Albin (2000) for smaller shifts, say 0.4σ or less, and the power of detecting larger shifts is the worst.

Table 15 and Figure 20 show the ARL performances for slope shifts. The EWMA chart is much better that the other charts over the entire range of shifts. Two metrics perform well for larger shifts with Absolute-value metric better. And the performance of EWMA₃ is worse than that for intercept shifts. So in this case, we can use the EWMA chart or EWMA₃ chart to detect the smaller shifts and use the EWMA chart or Absolute-value metric to detect the larger shifts of intercept and slope.

Table 16 and Figure 21 give the ARL values for standard deviation shifts. The

EWMSD chart also performs the best for smaller shifts, say 2.4 σ or less and EWMA/R chart and the R chart perform better than others for larger shifts. The detecting power of Squared metric, Absolute-value metric, and T² chart are similar. Our ARL comparisons show that our methods are slightly effective than the methods of Kang and Albin (2000) and Kim et al. (2003) for shifts in either the intercept or slope or increases in the error variance.

5. Conclusions

The profile monitoring is a very useful and promising area of research. In this thesis, we focus on the non-linear profile monitoring. Three profile monitoring approaches are proposed. And from the simulation studies, all approaches appear to perform well for the exponential profile under study.

Comparisons among these approaches are made. For I, M, and N shifts in the exponential profile, the EWMA chart performs the best over entire range of shifts, while the Absolute-value metric has secondly smaller ARL. And for standard deviation shifts, the EWMSD chart is a good choice to detect, while we also can use the R chart to monitor the process. And for linear profiles, our simulation study shows that new methods proposed by us are slightly effective than the methods of Kang and Albin (2000) and Kim et al. (2003). As a design issue, it is observed that increasing n is helpful in reducing the out-of-control ARL values. How to properly choose the number of bases is worth further studies.

This study extends the framework of statistical process control to more general applications. More statistical methods, models, and ideas are needed to extend the framework to a more complete profile monitoring strategy.

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Figure1. Milligrams of aspartame dissolved per liter of water from four samples.



Figure 2. 8 B-spline bases; each basis is of order 4.



Figure 4. The reference curve (solid line) and the curve affected by *I* shift (dotted line).



Figure 5. The reference curve (solid line) and the curve affected by *M* shift (dotted line).



Figure 6. The reference curve (solid line) and the curve affected by *N* shift (dotted line).



Figure 8. Scatter plot is the reference curve and the dashed line is the B-spline estimate.

Basis	1	2	3	4	5
Cc	2.874264	6.157306	11.611126	16.178782	16.183834
Z1-C	-0.0437711	0.0042971	0.00457416	0.00045094	-0.0013554
Z2-C	0.06077714	0.00684893	-0.0002403	-0.0109680	0.00645003
Z3-C	0.08942159	-0.0136533	0.01052427	-0.0160640	0.01881415
Basis	6	7	8	9	10
С	11.603308	6.176568	2.767467	1.423282	1.070967
Z1-C	-0.0031249	-0.0001978	0.00197017	-0.0013328	-0.0054912
Z2-C	-0.0079634	0.01199316	-0.0198097	0.0244964	-0.0223153
Z3-C	0.00019966	-0.0030393	0.00455043	0.01870481	-0.0077123
Basis	11	12	13		
U	1.008571	1.000276	1.003893		
Z1-U	0.00677231	0.00502447	-0.0936737		
Z2-U	0.00303271	0.01045864	-0.1506641		
Z3- U	-0.0079037	0.02057199	-0.2117598		

 Table 1. Simulation results to illustrate boundary effects; Underlines are larger deviations.

Cherry						α					
Cnart	0.00	0.05	0.10	0.15	0.20	0.25	0.30	0.35	0.40	0.45	0.50
\mathbf{T}^2	199.86	180.07	138.44	93.67	58.76	35.40	21.27	13.03	8.20	5.38	3.70
I	(0.622)	(0.569)	(0.437)	(0.294)	(0.183)	(0.110)	(0.066)	(0.040)	(0.024)	(0.015)	(0.010)
EWMA	200.42	47.39	14.71	7.62	5.06	3.79	3.06	2.59	2.27	2.04	1.87
	(0.613)	(0.135)	(0.033)	(0.014)	(0.007)	(0.005)	(0.003)	(0.003)	(0.002)	(0.002)	(0.002)
Squared metric	200.70	179.76	138.92	94.37	58.64	35.15	20.85	12.63	7.86	5.15	3.52
Squared metric	(0.622)	(0.566)	(0.435)	(0.297)	(0.184)	(0.109)	(0.064)	(0.038)	(0.023)	(0.015)	(0.009)
Absolute-value 1	199.49	175.20	134.78	88.79	53.08	30.87	17.76	10.53	6.49	4.23	2.93
metric	(0.625)	(0.561)	(0.424)	(0.277)	(0.167)	(0.096)	(0.055)	(0.032)	(0.019)	(0.012)	(0.008)

Table 2. ARL comparisons under I shifts from I_0 To $I_0 + \alpha \sigma$



Figure 9. ARL comparisons under I shifts of size α .

Cheert						β					
Cnart	0.0	0.1	0.2	0.3	0.4	0.5	0.6	0.7	0.8	0.9	1.0
\mathbf{T}^2	199.15	175.79	124.60	77.34	43.92	24.32	13.76	8.05	5.00	3.32	2.35
1	(0.623)	(0.560)	(0.391)	(0.243)	(0.137)	(0.075)	(0.042)	(0.024)	(0.014)	(0.009)	(0.006)
ЕТМИЛА	199.26	63.21	20.63	10.36	6.65	4.86	3.84	3.20	2.76	2.45	2.21
	(0.613)	(0.184)	(0.051)	(0.021)	(0.011)	(0.007)	(0.005)	(0.004)	(0.003)	(0.002)	(0.002)
Sauce de matrice	201.30	176.35	128.85	81.84	47.02	26.47	14.93	8.79	5.39	3.55	2.49
Squared metric	(0.623)	(0.554)	(0.405)	(0.257)	(0.147)	(0.082)	(0.046)	(0.026)	(0.015)	(0.010)	(0.006)
Absolute-value	199.37	173.90	122.30	74.54	42.39	23.79	13.56	8.14	5.10	3.43	2.45
metric	(0.620)	(0.548)	(0.385)	(0.233)	(0.132)	(0.074)	(0.041)	(0.024)	(0.014)	(0.009)	(0.006)

Table 3. ARL comparisons under M shifts from M_0 To $M_0 + \beta \sigma$





Cherry						γ					
Cnart	0.00	0.02	0.04	0.06	0.08	0.10	0.12	0.14	0.16	0.18	0.20
\mathbf{T}^2	199.4	173.86	120.61	73.36	41.99	23.82	13.80	8.38	5.37	3.67	2.66
ł	(0.628)	(0.547)	(0.380)	(0.231)	(0.131)	(0.073)	(0.042)	(0.025)	(0.015)	(0.010)	(0.007)
EWMA	201.3	50.39	16.21	8.44	5.61	4.23	3.43	2.91	2.55	2.30	2.11
	(0.608)	(0.144)	(0.038)	(0.016)	(0.009)	(0.006)	(0.004)	(0.003)	(0.002)	(0.002)	(0.002)
Squared metric	201.05	171.84	117.68	69.62	38.53	21.16	11.94	7.17	4.58	3.12	2.28
Squared metric	(0.638)	(0.542)	(0.369)	(0.219)	(0.121)	(0.065)	(0.036)	(0.021)	(0.013)	(0.008)	(0.005)
Absolute-value 1	199.35	169.74	114.29	66.73	36.79	20.34	11.67	7.14	4.63	3.19	2.36
metric	(0.621)	(0.537)	(0.361)	(0.210)	(0.115)	(0.063)	(0.035)	(0.021)	(0.013)	(0.008)	(0.006)

Table 4. ARL comparisons under N shifts from N_0 To $N_0 + \gamma \sigma$



Figure 11. ARL comparisons under N shifts of size γ .

Cheert						λ					
Chart	1.00	1.05	1.10	1.15	1.20	1.25	1.30	1.35	1.40	1.45	1.50
T^2	199.71	85.74	42.31	23.53	14.42	9.58	6.76	5.05	3.95	3.22	2.70
1	(0.624)	(0.270)	(0.132)	(0.073)	(0.044)	(0.029)	(0.020)	(0.014)	(0.011)	(0.008)	(0.007)
D	200.69	79.84	36.12	18.50	10.51	6.58	4.45	3.22	2.49	2.01	1.69
R	(0.634)	(0.253)	(0.113)	(0.057)	(0.032)	(0.019)	(0.012)	(0.008)	(0.006)	(0.005)	(0.003)
Squarad matria	200.23	81.03	38.73	21.03	12.58	8.26	5.80	4.33	3.41	2.78	2.34
Squareu metric	(0.632)	(0.255)	(0.121)	(0.065)	(0.038)	(0.025)	(0.017)	(0.012)	(0.009)	(0.007)	(0.006)
Absolute-value	199.62	85.42	42.48	23.68	14.39	9.53	6.70	5.03	3.91	3.17	2.65
metric	(0.628)	(0.269)	(0.133)	(0.073)	(0.044)	(0.029)	(0.020)	(0.014)	(0.011)	(0.008)	(0.007)
EWMAN	200.08	30.69	9.14	5.04	3.50	2.73	2.27	1.95	1.73	1.54	1.40
	(0.618)	(0.084)	(0.018)	(0.008)	(0.005)	(0.003)	(0.003)	(0.002)	(0.002)	(0.002)	(0.002)

Table 5. ARL comparisons under standard deviation shifts from σ To $\lambda\sigma$



Figure 12. ARL comparisons under standard deviation shifts of size λ

	T^2			β		
E Abso r	WMA lute-value netric	0.1	0.2	0.3	0.4	0.5
		132.527	86.171	50.568	28.459	16.038
		(0.417)	(0.272)	(0.158)	(0.088)	(0.049)
	-0.02	17.461	9.290	6.166	4.591	3.682
	-0.02	(0.042)	(0.018)	(0.010)	(0.006)	(0.004)
		127.463	82.180	46.314	26.716	14.851
		(0.401)	(0.259)	(0.146)	(0.083)	(0.045)
		81.769	51.071	29.896	17.262	10.149
		(0.257)	(0.160)	(0.093)	(0.053)	(0.030)
	_0.04	8.366	5.720	4.331	3.497	2.969
	-0.04	(0.016)	(0.009)	(0.006)	(0.004)	(0.003)
		75.389	45.172	26.194	15.078	8.938
		(0.236)	(0.141)	(0.081)	(0.045)	(0.027)
	-0.06	44.251	27.757	16.727	10.175	6.299
		(0.138)	(0.086)	(0.051)	(0.031)	(0.018)
		5.294	4.086	3.343	2.849	2.504
γ	-0.00	(0.008)	(0.005)	(0.004)	(0.003)	(0.002)
		39.093	23.681	14.163	8.560	5.328
		(0.124)	(0.074)	(0.043)	(0.026)	(0.015)
		22.900	14.721	9.376	5.992	4.021
		(0.070)	(0.045)	(0.028)	(0.017)	(0.011)
	0.08	3.856	3.187	2.745	2.421	2.185
	-0.00	(0.005)	(0.004)	(0.003)	(0.002)	(0.002)
		18.998	12.282	7.735	5.046	3.379
		(0.059)	(0.037)	(0.023)	(0.014)	(0.009)
		11.786	7.943	5.402	3.728	2.688
		(0.036)	(0.024)	(0.015)	(0.010)	(0.007)
		3 050	2 635	2 343	2 126	1 957
	-0.10	(0.003)	(0.003)	(0.002)	(0.002)	(0.002)
		10.052	6.615	4.420	2 111	2 200
		10.053	0.015	4.439	3.111	2.300
		(0.030)	(0.019)	(0.012)	(0.008)	(0.005)

Table 6. ARL comparisons under combinations of M and N shifts

	α									
lue	0.05	0.10	0.15	0.20	0.25					
	138.147	95.299	60.914	36.854	22.197					
	(0.436)	(0.301)	(0.190)	(0.115)	(0.068)					
1	17.308	8.387	5.375	3.961	3.166					
	(0.041)	(0.016)	(0.008)	(0.005)	(0.004)					
	132.874	90.184	55.410	32.321	19.042					
	(0.421)	(0.283)	(0.174)	(0.100)	(0.058)					
	89.390	58.511	36.237	22.241	13.733					
	(0.281)	(0.184)	(0.113)	(0.069)	(0.042)					
,	9.289	5.740	4.157	3.279	2.738					
-	(0.018)	(0.009)	(0.005)	(0.004)	(0.003)					
	85.664	53.523	32.597	18.822	11.481					
	(0.272)	(0.170)	(0.101)	(0.057)	(0.035)					
	52.159	33.431	20.896	13.239	8.467					
	(0.163)	(0.104)	(0.064)	(0.040)	(0.025)					
2	6.163	4.357	3.403	2.817	2.425					
,	(0.010)	(0.006)	(0.004)	(0.003)	(0.002)					
	49.722	31.011	19.080	11.463	7.294					
	(0.155)	(0.096)	(0.058)	(0.034)	(0.021)					
	29.134	18.945	12.252	8.056	5.420					
	(0.090)	(0.058)	(0.037)	(0.024)	(0.015)					
1	4.601	3.536	2.903	2.489	2.193					
• [(0.006)	(0.004)	(0.003)	(0.002)	(0.002)					
	27.553	17.439	11.081	7.096	4.615					
	(0.085)	(0.053)	(0.033)	(0.020)	(0.013)					
	16.393	10.946	7.410	5.129	3.637					
	(0.050)	(0.033)	(0.022)	(0.015)	(0.010)					
_	3 688	2 994	2 549	2 234	2 020					
5	(0.004)	(0.003)	(0.002)	(0.002)	(0.002)					
	15 675	10.151	6 600	1 560	3 201					
	(0.048)	(0.030)	(0.099	4.500	(0.008)					
	lue 1 2 3 4 5	Iue 0.05 138.147 (0.436) 17.308 (0.041) 132.874 (0.421) 89.390 (0.281) 9.289 (0.018) 85.664 (0.272) 52.159 (0.163) 3 6.163 (0.010) 49.722 (0.155) 29.134 (0.090) 4.601 (0.006) 27.553 (0.085) 16.393 (0.050) 3.688 (0.004) 15.675 (0.048) 15.675	Ine 0.05 0.10 138.147 95.299 (0.436) (0.301) 17.308 8.387 (0.041) (0.016) 132.874 90.184 (0.421) (0.283) 89.390 58.511 (0.281) (0.184) 9.289 5.740 (0.018) (0.009) 85.664 53.523 (0.272) (0.170) 52.159 33.431 (0.163) (0.104) 6.163 4.357 (0.010) (0.006) 49.722 31.011 (0.155) (0.096) 44.601 3.536 (0.006) (0.004) 27.553 17.439 (0.085) (0.053) 5 3.688 2.994 (0.004) (0.003) 15.675 10.151 (0.048) (0.030)	Ime 0.05 0.10 0.15 138.147 95.299 60.914 (0.436) (0.301) (0.190) 17.308 8.387 5.375 (0.041) (0.016) (0.008) 132.874 90.184 55.410 (0.421) (0.283) (0.174) 89.390 58.511 36.237 (0.281) (0.184) (0.113) 9.289 5.740 4.157 (0.018) (0.009) (0.005) 85.664 53.523 32.597 (0.272) (0.170) (0.101) 52.159 33.431 20.896 (0.163) 0.104) (0.064) 6.163 4.357 3.403 (0.010) (0.006) (0.004) (0.155) (0.096) (0.058) (0.155) (0.096) (0.037) 4 4.601 3.536 2.903 (0.006) (0.004) (0.003) (0.022) 5 3.688	Ine 0.05 0.10 0.15 0.20 138.147 95.299 60.914 36.854 (0.436) (0.301) (0.190) (0.115) 17.308 8.387 5.375 3.961 (0.041) (0.016) (0.008) (0.005) 132.874 90.184 55.410 32.321 (0.421) (0.283) (0.174) (0.100) 89.390 58.511 36.237 22.241 (0.281) (0.184) (0.113) (0.069) 9.289 5.740 4.157 3.279 (0.018) (0.009) (0.005) (0.004) 85.664 53.523 32.597 18.822 (0.272) (0.170) (0.101) (0.057) 52.159 33.431 20.896 13.239 (0.163) (0.104) (0.064) (0.040) 6.163 4.357 3.403 2.817 (0.010) (0.006) (0.004) (0.003) (0.155) (0.996)					

Table 7. ARL comparisons under combinations of I and M shifts.

	T ²			α		
E Abso r	WMA lute-value netric	0.05	0.10	0.15	0.20	0.25
		133.834	91.698	57.973	35.111	21.181
		(0.423)	(0.287)	(0.182)	(0.109)	(0.065)
	-0.02	15.043	7.713	5.088	3.808	3.073
	-0.02	(0.034)	(0.014)	(0.007)	(0.005)	(0.003)
		128.016	85.045	51.029	30.966	17.402
		(0.404)	(0.269)	(0.161)	(0.097)	(0.053)
		81.250	51.889	32.281	19.822	12.282
		(0.225)	(0.163)	(0.100)	(0.061)	(0.037)
	-0.04	7.702	5.074	3.802	3.062	2.596
	-0.04	(0.014)	(0.007)	(0.005)	(0.003)	(0.003)
		74.088	47.078	27.204	16.285	9.805
		(0.233)	(0.145)	(0.084)	(0.050)	(0.029)
		43.259	27.714	17.410	11.021	7.214
	-0.06	(0.135)	(0.086)	(0.053)	(0.033)	(0.021)
		5.013	3.773	3.045	2.582	2.265
γ		(0.007)	(0.005)	(0.003)	(0.003)	(0.002)
		37.935	23.307	14.090	8.931	5.658
		(0.119)	(0.070)	(0.044)	(0.027)	(0.016)
		22.217	14.523	9.459	6.309	4.375
		(0.069)	(0.044)	(0.028)	(0.018)	(0.012)
	-0.08	3.724	3.018	2.565	2.248	2.027
	-0.00	(0.005)	(0.003)	(0.002)	(0.002)	(0.002)
		18.990	12.084	7.822	5.085	3.490
		(0.059)	(0.036)	(0.023)	(0.015)	(0.009)
		11.466	7.803	5.389	3.822	2.824
		(0.035)	(0.023)	(0.015)	(0.010)	(0.007)
	0.40	2.967	2.531	2.222	2.010	1.841
	-0.10	(0.003)	(0.002)	(0.002)	(0.002)	(0.002)
		9 519	6 3 6 1	4 4 20	3 102	2 331
		(0.028)	(0.018)	(0.012)	(0.008)	(0.006)

Table 8. ARL comparisons under combinations of I and N shifts.

Chart	α										
Cnart	0.0	0.05	0.10	0.15	0.20	0.25	0.30	0.35	0.40	0.45	0.50
h - 13	199.6	180.28	138.93	94.34	58.84	35.13	20.89	12.68	7.88	5.14	3.53
<i>b</i> = 13	(0.626)	(0.569)	(0.441)	(0.297)	(0.184)	(0.109)	(0.064)	(0.038)	(0.023)	(0.015)	(0.009)
h = 0	199.22	176.11	128.00	81.01	47.56	27.31	15.80	9.46	5.93	3.92	2.77
<i>v</i> – <i>y</i>	(0.626)	(0.556)	(0.403)	(0.225)	(0.147)	(0.085)	(0.048)	(0.028)	(0.017)	(0.011)	(0.007)
h – 5	199.61	168.45	108.79	61.25	33.08	17.98	10.26	6.19	4.01	2.76	2.04
v = 5	(0.629)	(0.529)	(0.345)	(0.193)	(0.103)	(0.055)	(0.031)	(0.018)	(0.011)	(0.007)	(0.005)

Table 9. ARL comparisons of T^2 chart with different number of bases for *I* shifts.



Figure 13. ARL comparisons of T^2 chart with different number of bases for *I* shifts.

Cheert		α										
Cnart	0.00	0.05	0.10	0.15	0.20	0.25	0.30	0.35	0.40	0.45	0.50	
n - 50	199.97	180.01	139.09	94.53	58.39	35.19	20.97	12.62	7.88	5.16	3.51	
n = 50	(0.624)	(0.567)	(0.438)	(0.296)	(0.182)	(0.110)	(0.064)	(0.038)	(0.023)	(0.015)	(0.009)	
n = 40	199.30	184.07	148.98	107.22	72.16	46.37	29.02	18.24	11.73	7.67	5.26	
n = 40	(0.621)	(0.579)	(0.470)	(0.338)	(0.226)	(0.145)	(0.090)	(0.056)	(0.035)	(0.023)	(0.015)	
n = 20	199.1	187.94	159.24	124.77	90.39	62.39	42.30	28.28	18.86	12.78	8.88	
<i>n</i> = 30	(0.622)	(0.593)	(0.503)	(0.393)	(0.284)	(0.196)	(0.133)	(0.088)	(0.058)	(0.039)	(0.026)	

Table 10. ARL comparisons of T^2 chart with different *n*'s for *I* shifts.



Figure 14. ARL comparisons of T^2 chart with different *n*'s for *I* shifts.



Figure 15. The reference curve (solid line) and the fitted B-spline curve with 5 bases and n=50.

Chart		β											
	0.0	0.1	0.2	0.3	0.4	0.5	0.6	0.7	0.8	0.9	1.0		
	201.13	63.98	20.69	10.43	6.66	4.87	3.86	3.21	2.77	2.45	2.22		
n = 50	(0.627)	(0.186)	(0.051)	(0.021)	(0.011)	(0.007)	(0.005)	(0.004)	(0.003)	(0.002)	(0.002)		
	199.86	73.48	24.94	12.45	7.82	5.67	4.43	3.65	3.12	2.74	2.46		
n = 40	(0.615)	(0.236)	(0.064)	(0.027)	(0.014)	(0.009)	(0.006)	(0.004)	(0.003)	(0.003)	(0.002)		
n = 30	200.23	88.39	32.19	15.88	9.83	6.95	5.35	4.36	3.68	3.21	2.85		
n=30	(0.628)	(0.264)	(0.087)	(0.037)	(0.019)	(0.012)	(0.008)	(0.006)	(0.004)	(0.004)	(0.003)		
<i>n</i> =20	201.60	109.19	44.66	22.57	13.69	9.45	7.10	5.69	4.73	4.07	3.58		
	(0.637)	(0.329)	(0.125)	(0.057)	(0.030)	(0.018)	(0.012)	(0.009)	(0.007)	(0.005)	(0.004)		

Table 11. ARL comparisons of EWMA chart with different n's for M shifts.



Figure 16. ARL comparisons of the EWMA chart with different *n*'s for *M* shifts.

Chart		λ											
	1.00	1.05	1.10	1.15	1.20	1.25	1.30	1.35	1.40	1.45	1.50		
<i>n</i> = 50	200.36	81.03	38.73	21.03	12.58	8.26	5.83	4.35	3.41	2.77	2.32		
	(0.632)	(0.255)	(0.121)	(0.065)	(0.038)	(0.025)	(0.017)	(0.012)	(0.009)	(0.007)	(0.005)		
	200.22	89.38	44.15	26.48	17.32	11.64	8.22	5.58	4.17	3.36	2.71		
n = 40	(0.639)	(0.272)	(0.142)	(0.079)	(0.045)	(0.033)	(0.023)	(0.016)	(0.011)	(0.008)	(0.006)		
n - 30	200.68	96.10	52.55	32.67	22.17	15.11	10.59	6.63	5.14	3.75	3.24		
n = 30	(0.631)	(0.296)	(0.169)	(0.090)	(0.057)	(0.039)	(0.031)	(0.021)	(0.015)	(0.010)	(0.007)		
<i>n</i> =20	201.09	106.97	62.29	40.09	30.61	19.98	14.77	9.40	6.49	4.45	4.82		
	(0.643)	(0.324)	(0.194)	(0.112)	(0.083)	(0.048)	(0.040)	(0.029)	(0.020)	(0.013)	(0.009)		

 Table 12. ARL comparisons of Squared metric with different *n*'s for standard deviation shifts.



Figure 17. ARL comparisons of Squared metric with different *n*'s for standard deviation shifts.

Chart		α											
	0.00	0.05	0.10	0.15	0.20	0.25	0.30	0.35	0.40	0.45	0.50		
n - 50	200.25	176.64	134.26	88.90	54.20	30.86	17.93	10.59	6.58	4.22	2.96		
n = 50	(0.631)	(0.559)	(0.423)	(0.280)	(0.169)	(0.097)	(0.056)	(0.032)	(0.019)	(0.011)	(0.007)		
<i>n</i> – 40	200.46	179.69	145.41	103.91	66.33	42.20	25.16	15.90	9.83	6.45	4.29		
n = 40	(0.632)	(0.582)	(0.457)	(0.328)	(0.210)	(0.133)	(0.076)	(0.049)	(0.029)	(0.019)	(0.012)		
n = 20	201.73	184.97	156.89	123.00	85.88	59.14	38.39	25.31	16.33	11.00	7.45		
n=30	(0.642)	(0.591)	(0.494)	(0.387)	(0.268)	(0.186)	(0.120)	(0.077)	(0.050)	(0.033)	(0.022)		
20	200.42	190.15	174.07	142.36	110.42	85.34	62.67	42.88	31.71	22.46	15.96		
n = 20	(0.632)	(0.598)	(0.556)	(0.444)	(0.346)	(0.272)	(0.197)	(0.134)	(0.098)	(0.069)	(0.049)		

Table 13. ARL comparisons of Absolute-value metric with different n's for I shifts



Figure 18. ARL comparisons of Absolute-value metric with different *n*'s for *I* shifts.

		λ												
Cnart	0.0	0.2	0.4	0.6	0.8	1.0	1.2	1.4	1.6	1.8	2.0			
EWMA/R	198.85	65.58	17.54	8.32	5.33	3.95	3.17	2.67	2.33	2.10	1.94			
	(0.619)	(0.192)	(0.040)	(0.013)	(0.007)	(0.004)	(0.003)	(0.002)	(0.002)	(0.001)	(0.001)			
\mathbf{T}^2	198.44	137.21	64.0	27.96	13.20	6.90	4.00	2.57	1.84	1.45	1.23			
1	(0.627)	(0.433)	(0.198)	(0.087)	(0.040)	(0.020)	(0.011)	(0.006)	(0.004)	(0.003)	(0.002)			
FWMA.	199.36	60.23	17.28	8.91	6.12	4.82	4.08	3.59	3.27	3.05	2.89			
	(0.614)	(0.170)	(0.036)	(0.013)	(0.007)	(0.004)	(0.003)	(0.002)	(0.002)	(0.001)	(0.001)			
FWMA	200.6	39.28	12.02	6.37	4.30	3.28	2.68	2.29	2.03	1.84	1.67			
	(0.616)	(0.109)	(0.025)	(0.010)	(0.006)	(0.004)	(0.003)	(0.002)	(0.002)	(0.002)	(0.002)			
Squared metric	201.43	137.40	63.31	27.90	13.17	6.89	3.99	2.58	1.84	1.45	1.23			
Squarea mente	(0.638)	(0.435)	(0.199)	(0.087)	(0.040)	(0.020)	(0.011)	(0.006)	(0.004)	(0.003)	(0.002)			
Absolute-value	199.92	124.64	52.03	21.91	10.31	5.45	3.25	2.19	1.63	1.33	1.16			
metric	(0.627)	(0.395)	(0.163)	(0.067)	(0.031)	(0.016)	(0.009)	(0.005)	(0.003)	(0.002)	(0.001)			

Table 14. ARL comparisons under intercept shifts of a linear profile.





Figure 19. ARL comparisons under intercept shifts of a linear profile.

						β					
Cnart	0.000	0.025	0.050	0.075	0.100	0.125	0.150	0.175	0.200	0.225	0.250
EWMA/R	198.75	117.11	43.56	19.67	11.33	7.71	5.80	4.65	3.89	3.46	2.97
	(0.625)	(0.361)	(0.122)	(0.047)	(0.022)	(0.013)	(0.008)	(0.006)	(0.005)	(0.004)	(0.003)
\mathbf{T}^2	200.25	165.33	105.43	60.30	34.31	20.14	12.18	7.81	5.23	3.70	2.74
-	(0.630)	(0.523)	(0.322)	(0.191)	(0.107)	(0.062)	(0.037)	(0.023)	(0.015)	(0.010)	(0.007)
FWMA.	200.3	102.65	37.33	18.12	11.22	8.14	6.47	5.45	4.75	4.27	3.90
	(0.614)	(0.306)	(0.098)	(0.038)	(0.019)	(0.011)	(0.007)	(0.005)	(0.004)	(0.003)	(0.003)
ЕМЛА	200.79	77.93	26.83	13.41	8.39	6.00	4.69	3.85	3.28	2.87	2.57
	(0.622)	(0.232)	(0.070)	(0.030)	(0.016)	(0.009)	(0.007)	(0.005)	(0.004)	(0.003)	(0.002)
Squared metric	199.50	165.12	105.32	60.74	34.59	20.19	12.32	7.83	5.20	3.71	2.74
Squarea meare	(0.627)	(0.524)	(0.330)	(0.190)	(0.108)	(0.062)	(0.037)	(0.023)	(0.015)	(0.010)	(0.007)
Absolute-value	200.71	159.54	96.56	54.59	30.90	18.14	11.27	7.28	4.95	3.42	2.67
metric	(0.623)	(0.504)	(0.304)	(0.171)	(0.096)	(0.056)	(0.034)	(0.021)	(0.014)	(0.009)	(0.007)

Table 15. ARL comparisons under slope shifts of a linear profile.



Figure 20. ARL comparisons under slope shifts of a linear profile

						γ					
Cnart	1.0	1.2	1.4	1.6	1.8	2.0	2.2	2.4	2.6	2.8	3.0
EWMA/R	198.55	34.09	12.07	6.15	3.96	2.86	2.28	1.93	1.70	1.53	1.42
	(0.623)	(0.105)	(0.036)	(0.017)	(0.010)	(0.007)	(0.005)	(0.004)	(0.003)	(0.002)	(0.002)
\mathbf{T}^2	200.32	39.73	15.01	7.91	5.14	3.76	2.98	2.51	2.20	1.96	1.80
1	(0.633)	(0.123)	(0.046)	(0.023)	(0.015)	(0.010)	(0.008)	(0.006)	(0.005)	(0.004)	(0.004)
FWMA.	199.2	34.51	13.80	8.31	6.10	4.97	4.27	3.84	3.51	3.26	3.07
	(0.606)	(0.095)	(0.032)	(0.016)	(0.010)	(0.007)	(0.005)	(0.004)	(0.004)	(0.003)	(0.003)
R	200.1	34.38	12.12	6.20	3.99	2.89	2.31	1.96	1.73	1.57	1.46
K	(0.638)	(0.107)	(0.037)	(0.018)	(0.011)	(0.007)	(0.006)	(0.004)	(0.004)	(0.003)	(0.003)
Squared metric	201.27	39.54	14.92	7.92	5.14	3.74	2.99	2.52	2.19	1.96	1.80
Squarea metric	(0.638)	(0.123)	(0.045)	(0.024)	(0.015)	(0.010)	(0.008)	(0.006)	(0.005)	(0.004)	(0.004)
Absolute-value	199.92	41.90	15.99	8.45	5.43	3.93	3.13	2.62	2.26	2.02	1.85
metric	(0.632)	(0.131)	(0.049)	(0.025)	(0.015)	(0.011)	(0.008)	(0.006)	(0.005)	(0.004)	(0.004)
FWMV	199.83	20.19	7.46	4.44	3.22	2.57	2.17	1.91	1.73	1.59	1.49
E VV IVI V	(0.629)	(0.058)	(0.019)	(0.010)	(0.007)	(0.005)	(0.004)	(0.003)	(0.003)	(0.003)	(0.002)

Table 16. ARL comparisons under standad deviation shifts of a linear profile.





Figure 21. ARL comparisons under standad deviation shifts of a linear profile.