

Joint Frequency and DOA Estimation with Automatic Pairing Using the Rayleigh–Ritz Theorem

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Abstract: This paper presents a novel scheme for joint frequency and direction of arrival (DOA) estimation, that pairs frequencies and DOAs automatically without additional computations. First, when the property of the Kronecker product is used in the received array signal of the multiple-delay output model, the frequency-angle steering vector can be reconstructed as the product of the frequency steering vector and the angle steering vector. The frequency of the incoming signal is then obtained by searching for the minimal eigenvalue among the smallest eigenvalues that depend on the frequency parameters but are irrelevant to the DOAs. Subsequently, the DOA related to the selected frequency is acquired through some operations on the minimal eigenvector according to the Rayleigh–Ritz theorem, which realizes the natural pairing of frequencies and DOAs. Furthermore, the proposed method can not only distinguish multiple sources, but also effectively deal with other arrays. The effectiveness and superiority of the proposed algorithm are further analyzed by simulations.

Keywords: Array signal processing; DOA estimation; automatic pairing; Rayleigh-Ritz theorem

1 Introduction

Frequency estimation and DOA estimation for an antenna array are the basic but key problems in the area of array signal processing [1–3], e.g., in the applications of direction finding, radar, sonar, wireless communications, and especially location service [4,5] under the scenario of multiple incoming signals with different frequencies. The maximum likelihood (ML) method [6] is often applied but brings a great computational load. To reduce the computational effort required, the multiple signal classification (MUSIC) algorithm [7], some ESPRIT-based joint angle and frequency estimation methods [8,9] and the propagator method (PM) [10] have been proposed. However, the MUSIC method [7] requires spectral peak searching in both the frequency range and angle range, and this is still computationally expensive, particularly for 2-dimensional arrays. The PM [10] has low complexity, but its parameter estimation performance is poorer than that of the ESPRIT algorithm. Unfortunately, ESPRIT is only applicable to a uniform linear array.



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In [11], the iterative least-squares method was used for joint frequency and DOA estimation with L-shaped arrays, and many iterations were needed for convergence.

Using multiple-delay outputs, the authors in [12] proposed a joint angle and frequency estimation method based on ESPRIT; this approach outperforms the conventional ESPRIT algorithm, but it not only requires additional pairing operations [13] but is also only suitable for uniform linear arrays. In addition, the angle estimation procedure only utilizes part of the received signals. The ESES method proposed in a recent study [14] has the same procedure for estimating frequency as the method in [12], but unlike in [12], all of the received signals are used to estimate the angle in [14]; thus, the performance is better than the method developed in [12]. The ESES approach [14] can also be employed for other array structures. However, the ESES method in [14] carelessly neglects that the eigenvalues of two matrices from the different permutations of one matrix may not be in the same order, so the frequencies and angles obtained from the eigenvalues require some additional computations for pairing those parameters (as in [13]), and the process cannot always arrive at the correct pairings under some conditions.

To pair the frequency and angle estimations automatically and avoid potential pairing mistakes, relative to the multiple-delay output model used in [12,14], this paper is an improvement and presents a novel joint frequency and angle estimation algorithm without additional pairings owing to the Rayleigh–Ritz theorem [15, Section 4.2]. Frequencies can be calculated by a one-dimensional search, and angles can be obtained as closed-form solutions. Furthermore, the proposed algorithm can work with other arrays, such as nonuniform linear arrays. Hence, a unified framework for joint frequency and angle estimation performance than that of the ESES method in [14] and has acceptable computational complexity, which is confirmed by simulation results.

An outline of this paper is as follows. Section 2 describes the data model. Section 3 addresses the algorithmic issues encountered with some remarks. Section 4 shows the simulation results, which are followed by conclusions in Section 5.

Notations: $(\cdot)^T$, $(\cdot)^H$, and $(\cdot)^{\dagger}$ denote the transpose, conjugate transpose and pseudoinverse operations, respectively; diag(v) stands for a diagonal matrix whose *i*th diagonal element is the *i*th entry in vector v; I_M is an $M \times M$ identity matrix; \otimes represents the Kronecker product; $Rank(\cdot)$ is the rank of the chosen matrix; $angle(\cdot)$ returns the phase angle of the chosen element; and arg max and arg min stand for the maximum and minimum arguments, respectively.

2 Data Model

Consider an arbitrary array of M sensors on which K (assuming K is known in this study) incident waves impinge; if the sources are all narrow-band signals without the same center frequency, the signal received at the *m*th antenna can be expressed as

$$x_m(t) = \sum_{k=1}^{K} e^{j2\pi f_k \tau_{m,k}} s_k(t) + n_m(t)$$

$$m = 1, 2, \dots, M, \quad k = 1, 2, \dots, K,$$
(1)

where f_k , $s_k(t)$ are the frequency and narrow-band signal of the kth source, respectively. $n_m(t)$ represents spatially and temporally additive Gaussian white noise with zero mean. $\tau_{m,k}$ is the time

delay of the *m*th sensor; therefore, it depends on the structure of the array and the *k*th incoming angle of the received signal. Thus, $\tau_{m,k}$ of the uniform linear array can be denoted by

$$\tau_{m,k} = \frac{(m-1)d}{c}\cos\theta_k, \quad m = 1, 2, \dots, M, \quad k = 1, 2, \dots, K,$$
(2)

where d is the distance between the adjoining sensors, c is the velocity of light, and θ_k is the incoming angle of the kth source. The output of the array is

$$X_0 = [x_1(t), x_2(t), \dots, x_M(t)]^T.$$
(3)

The received signal of array antennas can be shown in matrix form as

$$X_0 = AS + N_0, \tag{4}$$

where

$$S = [s_1, s_1, \dots, s_K]^T \in C^{K \times N},$$

$$N_0 = [n_1, n_1, \dots, n_M]^T \in C^{M \times N},$$

$$A = [a_1, a_2, \dots, a_K]$$

$$= \begin{bmatrix} e^{j2\pi f_k \tau_{1,1}} & e^{j2\pi f_k \tau_{1,2}} & \dots & e^{j2\pi f_k \tau_{1,K}} \\ e^{j2\pi f_k \tau_{2,1}} & e^{j2\pi f_k \tau_{2,2}} & \dots & e^{j2\pi f_k \tau_{2,K}} \\ \vdots & \vdots & \ddots & \vdots \\ e^{j2\pi f_k \tau_{M,1}} & e^{j2\pi f_k \tau_{M,2}} & \dots & e^{j2\pi f_k \tau_{M,K}} \end{bmatrix}$$

The angle steering vector is

$$a_{k}(\theta) = \left[e^{2\pi f_{k}\tau_{1,k}}, e^{2\pi f_{k}\tau_{2,k}}, \dots, e^{2\pi f_{k}\tau_{M,k}}\right]^{T},$$

and the delayed signal for (1) with δ can be written as

$$x_{m}(t-\delta) = \sum_{k=1}^{K} e^{j2\pi f_{k}\tau_{m,k}} s_{k}(t-\delta) + n'_{m}(t)$$

$$= \sum_{k=1}^{K} e^{j2\pi f_{k}\tau_{m,k}} s_{k}(t) e^{-j2\pi f_{k}\delta} + n'_{m}(t),$$

$$m = 1, 2, ..., M.$$
 (5)

Afterwards, the delayed signal for (5) with δ can be denoted as $X_1 = A \Upsilon S + N_1$, where $\Upsilon = diag[\omega_1, \omega_2, ..., \omega_K], \ \omega_K = e^{-j2\pi f_k \delta}, \ k = 1, 2, ..., K$.

(6)

Therefore, the delayed signal for (1) with $p\delta$ can be expressed as

$$X_p = A\Upsilon^p S + N_p, \quad p = 0, 1, 2, \dots, P - 1,$$
where P is the number of delays.
(7)

According to (4), (6) and (7), the multiple-delay output model [12,14] can be described as

$$X = \begin{bmatrix} X_{0} \\ X_{1} \\ \vdots \\ X_{P-1} \end{bmatrix} = \begin{bmatrix} A \\ A\Upsilon \\ \vdots \\ A\Upsilon^{p-1} \end{bmatrix} S + \begin{bmatrix} N_{0} \\ N_{1} \\ \vdots \\ N_{P-1} \end{bmatrix}$$

$$= \begin{bmatrix} a_{1} & a_{2} & \cdots & a_{K} \\ \omega_{1}a_{1} & \omega_{1}a_{1} & \cdots & \omega_{K}a_{K} \\ \vdots & \vdots & \ddots & \vdots \\ \omega_{1}^{P-1}a_{1} & \omega_{2}^{P-1}a_{2} & \cdots & \omega_{K}^{P-1}a_{K} \end{bmatrix} S + \begin{bmatrix} N_{0} \\ N_{1} \\ \vdots \\ N_{P-1} \end{bmatrix}.$$
(8)

Define the frequency steering vector $\beta_k(f) = \left[1, \omega_k, \omega_k^2, \omega_k^3, \dots, \omega_k^{P-1}\right]^T$, $k = 1, 2, \dots, K$. Then, (8) can be reconstructed as

$$X = [\beta_1(f) \otimes a_1(\theta) \quad \beta_2(f) \otimes a_2(\theta) \quad \cdots \quad \beta_K(f) \otimes a_K(\theta)]S + N,$$
(9)

where $\beta(f) \otimes a(\theta)$ is the frequency-angle steering vector.

3 Joint Frequency and Angle Estimation

From (9), we know that the frequency f_k and angle θ_k can be estimated by attaining β_k and a_k from the received array signal. The covariance matrix of the received signal can be calculated via $R_{xx} = XX^H$. Using the eigenvalue decomposition of R_{xx} , we can obtain E_n , which is the noise space spanned by the (MP - K) eigenvectors corresponding to the minimal (MP - K) eigenvalues of the covariance matrix R_{xx} . Hence, the frequency f_k and angle θ_k can be obtained by the MUSIC algorithm [1,7] as

$$(f_k, \theta_k) = \underset{(f,\theta)}{\operatorname{arg\,max}} S(f, \theta)$$

$$= \underset{(f,\theta)}{\operatorname{arg\,max}} \frac{1}{(\beta(f) \otimes a(\theta))^H E_n E_n^H (\beta(f) \otimes a(\theta))},$$

$$k = 1, 2, \dots, K.$$
(10)

However, to obtain all source parameters (f_k, θ_k) from (10), an exhaustive search in the two-dimensional space is generally required, thereby resulting in a heavy computational burden. Here, by the property of the Kronecker product, a novel implementation of the two-dimensional parameter estimation process without the exhaustive two-dimensional search is developed based on the Rayleigh–Ritz theorem [15, Section 4.5].

3.1 Method 1: EVEV

By the property of the Kronecker product, the frequency-angle steering vector can be reformulated in the following form

$$\beta(f) \otimes a(\theta) = \beta(f) \otimes I_M a(\theta).$$
⁽¹¹⁾

From (11), it is obvious that the frequency-angle steering vector is decomposed into the product of one matrix $(\beta(f) \otimes I_M)$ that is only related to the frequency steering vector and the other matrix $(a(\theta))$ that is only concerned with the angle steering vector; this product lays a foundation for estimating f_k and θ_k separately and enables us to avoid a two-dimensional search.

Inserting (11) into (10), we can obtain

$$(f_k, \theta_k) = \underset{(f,\theta)}{\operatorname{arg\,max}} \frac{1}{a(\theta)^H \left(\beta(f) \otimes I_M\right)^H E_n E_n^H \left(\beta(f) \otimes I_M\right) a(\theta)}, \quad k = 1, 2, \dots, K$$
(12)

Define

...

$$T(f) = (\beta(f) \otimes I_M)^H E_n E_n^H (\beta(f) \otimes I_M), \qquad (13)$$

where T(f) is an $M \times M$ Hermitian matrix dependent on f and independent of θ . Thus, (12) can be rewritten as

$$(f_k, \theta_k) = \underset{(f,\theta)}{\operatorname{arg\,max}} \frac{1}{a(\theta)^H T(f) a(\theta)}$$

$$= \underset{(f,\theta)}{\operatorname{arg\,min}} a(\theta)^H T(f) a(\theta)$$
(14)

Since T(f) is a Hermitian matrix, the Rayleigh quotient of T(f) is a scalar that can be defined as

$$Q(f,\theta) = \frac{a(\theta)^{H} T(f) a(\theta)}{a(\theta)^{H} a(\theta)}$$
(15)

Notice that $a(\theta) \neq 0$ and $a(\theta)^H a(\theta) \equiv M$; thus, (14) can be converted into a Rayleigh quotient problem:

$$(f_k, \theta_k) = \underset{(f,\theta)}{\operatorname{arg\,max}} \frac{1}{a(\theta)^H T(f) a(\theta)}$$

$$= \underset{(f,\theta)}{\operatorname{arg\,min}} a(\theta)^H T(f) a(\theta) \tag{16}$$

$$= \underset{(f,\theta)}{\operatorname{arg\,min}} Q(f,\theta) = \underset{(f,\theta)}{\operatorname{arg\,min}} \frac{a(\theta)^H T(f) a(\theta)}{a(\theta)^H a(\theta)}$$

If f does not correspond to one of the true values, and if

$$Rank = \left(E_n E_n^H\right) = MP - K \ge M \Leftrightarrow K \le M \left(P - 1\right),\tag{17}$$

then, T(f) is invertible, and its eigenvalues are all greater than zero. According to the Rayleigh-Ritz theorem in [15], the minimum value of the objective in (16) is the minimum eigenvalue of T(f), and in this case, $a(\theta)$ should be proportional to the minimum eigenvector of T(f); thus, the frequency f_k satisfies

$$f_k = \underset{f}{\arg\min} \rho_{\min}(T(f)), \quad k = 1, 2, ..., K,$$
 (18)

and θ_k has the following relationship with $T(f_k)$:

$$a(\theta_k) = he_{\min}(T(f_k)), \quad k = 1, 2, \dots, K,$$
(19)

where $\rho_{\min}(T(f))$ is the minimal eigenvalue of the matrix T(f), $e_{\min}(T(f_k))$ is the eigenvector corresponding to the minimal eigenvalue of the matrix $T(f_k)$, and h is a constant. There are many fast algorithms, such as [16], for finding the minimal eigenvalue and eigenvector of a Hermitian matrix.

To solve for θ_k from (19), let $\xi_m(f_k)$ denote the *k*th phase angle of the *m*th element of the eigenvector $e_{\min}(T(f_k))$, that is, $\xi_m(f_k) = angle(e_{\min,m}(T(f_k)))(\xi_m(f_k) \in (-\pi, \pi])$, where $e_{\min,m}(T(f_k))$ is the *m*th element of $e_{\min}(T(f_k))$. Then, it follows that

$$2\pi f_k \left(\tau_{q+1}, k - \tau_{q,k}\right) = \xi_{q+1} \left(f_k\right) - \xi_q \left(f_k\right) + 2k_q \pi,$$

$$k_q \in \{0, \pm 1, \pm 2, \ldots\}, \quad q = 1, 2, \ldots, M - 1,$$
(20)

in which $(\xi_{q+1}(f_k) - \xi_q(f_k)) \in (-\pi, \pi].$

For an unambiguous uniform linear array, the interspacing between two adjacent elements d is not greater than $\lambda/2$ (λ is the wavelength of the signal), so $-\pi < 2\pi f_k (\tau_{q+1}, k - \tau_{q,k}) \le \pi$ and $k_q = 0$. Afterwards, recall the f_k obtained by (18), and the accurate value of θ_k can be estimated by solving the equation set of (20) using the least-squares method.

It should be pointed out that θ_k is obtained by inserting f_k into (19) and solving the equation set of (20); thus, it is inferred that f_k and θ_k are paired naturally.

The proposed method can also address the case in which multiple signals have the same frequency but different DOAs by distinguishing the number of minimal eigenvalues with respect to f_k in (18) and their corresponding eigenvectors in (19).

Because the frequency is obtained via an eigenvalue and the angle is obtained via an eigenvector, this method is named the EVEV algorithm.

3.2 Method 1: ESEV

It is worth noting that to obtain the frequency f_k from (18), a one-dimensional search in the frequency space is required. If the true frequency lies in a set with few values, there is little computational expense. In contrast, if the estimated frequency depends on a wide range of frequencies, this may lead to a great computational load. To pursue a more convenient implementation of joint frequency and angle estimation than the EVEV algorithm, the DOA θ_k can be found from (20) subsequent to the frequency f_k estimated by the ESPRIT algorithm, as in [12,14].

By the eigenvalue decomposition of R_{xx} , the signal subspace E_s is spanned by the K eigenvectors corresponding to the first K maximal eigenvalues. Then, we divide E_s into E_1 (the first M(P-1) rows of E_s) and E_2 (the last M(P-1) rows of E_s). Define

$$\Psi_{K \times K} = E_1^{\dagger} E_2 \tag{21}$$

Then we can obtain the eigenvalues ρ_k (k = 1, 2, ..., K) of the matrix Ψ . Finally, the frequency f_k is estimated by (see [12,14] for more information)

$$f_k = \frac{1}{2\pi\delta} angle(\rho_k) \quad k = 1, 2, ..., K$$
 (22)

Subsequently, based on the Rayleigh-Ritz theorem, T(f) is constructed with (13), and then the DOA is obtained by substituting the f_k obtained from (22) into (20), as in the previous subsection. This method, which is named the ESEV algorithm, is a combination of the ESES method in [14] and the EVEV method proposed in this study, and it provides joint frequency and angle estimations in closed-form solutions.

3.3 Complexity Analysis

In this section, we analyze the computational complexities of the EVEV and ESEV algorithms and compare them with that of the ESES method [14]. For the EVEV algorithm, obtaining the covariance matrix estimation R_{xx} requires $O(LM^2P^2)$ flops, obtaining E_n through the eigendecomposition of R_{xx} costs $O(M^3P^3)$ flops, and the estimation of T(f) and obtaining the minimal eigenvalue of T(f) and its eigenvector via several iterative methods requires $O(M^2P^2(MP-K) + 2r(M^2P^2 + M^2))$ flops. Thus, the total computational complexity of the EVEV algorithm is $O\{LM^2P^2 + M^3P^3 + M^2P^2(MP-K) + r(2M^2P^2 + M^2)\}$ flops, where r is the number of search grids within the search frequency region. To pair the frequency and DOA, the ESES method needs $O(2K^3 + K^2)$ flops. Hence, the ESES algorithm requires $O\{LM^2P^2 + M^3P^3 + [2K^2M(P-1) + 2K^3 + K^3] + [2K^2(M-1)P + 2K^3 + 2K^3] + (2K^3 + K^2)\}$ flops in total. Therefore, the proposed ESEV algorithm needs $O\{LM^2P^2 + M^3P^3 + [2K^2M(P-1) + 2K^3 + K^3] + M^2P^2(MP-K) + K(2M^2P^2 + M^2)\}$ flops.

It can be concluded that the EVEV method has the largest computational complexity among these three methods if the number of search grids r is large. Furthermore, both the ESEV and ESES methods provide frequency and DOA estimations in closed-form solutions, as mentioned previously, so their computational complexities with respect to r are irrelevant. In addition, the ESEV method has almost the same computational complexity as that of the ESES method.

3.4 Remarks

From these aforementioned derivations, some remarks can be made:

Remark 1: From (17), the number of resolvable sources satisfies $K \le M (P-1)$; however, (18) indicates that K must not be greater than the dimension of the matrix T(f), that is, $K \le M$. In general, the number of delays satisfies $P \ge 2$, so the maximum number of resolvable sources satisfies $K \le M$ for the EVEV method. The ESES method only estimates the frequency and DOA if $K \le M$; therefore, the maximum numbers of resolvable sources for the ESES and ESEV methods also satisfy $K \le M$. The abilities of these three methods to resolve multiple sources, based on the multiple-delay output model, exceed those of conventional algorithms [6–11], which can resolve at most (M-1) sources.

Remark 2: Both the EVEV algorithm and ESEV algorithm can achieve the automatic pairing of the estimation parameters. The angle θ_k is obtained by inserting f_k into (19); thus, the frequency f_k and direction of arrival θ_k can be paired naturally with both the EVEV and ESEV methods.

Remark 3: Both the EVEV algorithm and ESEV algorithm can be applied to arbitrary arrays. Obtaining the frequency from (18) or (22) is irrelevant to the array geometry, and angle estimation can be realized from (20) based on prior information about the array structure. Therefore, the two proposed methods have no dependencies on the array configurations and can be used with any array, for example, a nonuniform linear array.

Remark 4: From the delay segment $e^{j2\pi f_k\delta}$ in (5), it is easy to see that the delay time must satisfy $0 < \delta < 1/\max(f_k)$ to avoid ambiguity in the estimation of the frequency f_k ; this is a milder requirement for δ than $0 < (P-1)\delta < 1/\max(f_k)$ in [12].

Remark 5: The EVEV, ESEV and ESES methods can all deal with the case in which two or more signals have the same frequency but are from different incoming DOAs. In contrast, none of these methods can work effectively in a coherent source environment.

4 Numerical Results

The performances of the proposed methods are evaluated with some Monte Carlo simulations. In the following experiments, assume that M is the number of antennas, P is the number of delays, L is the number of snapshots, and K is the number of sources. The distance between two adjacent elements in the uniform linear array is half of the wavelength of the maximum estimated frequency, and the other parameters are M = 10, P = 3, L = 200, signal-to-noise ratio $SNR = 20 \ dB$, sampling frequency $f_s = 2000000 \ H_Z$ and delay time $\delta = 1/(f_s/2)$. The following results are evaluated by the estimated root mean square errors (RMSEs) obtained from the average results of 2000 Monte Carlo simulations. Frequency RMSE stands for the RMSE of the estimated frequencies, and DOARMSE stands for the RMSE of the estimated DOAs. If there is no additional instruction, these parameters all hold in the simulations. In addition, we compare EVEV and ESEV with ESES [14], which has the same data model as the one in this study.

Simulation 1. The performance of ESES is investigated. Tab. 1 indicates the results of the ESES, EVEV and ESEV methods when the angles are 20° , 50° and 80° , and their carrier frequencies are 200000, 400000 and 800000 H_Z , respectively. From Tab. 1, it is easy to see that the ESES method has estimation ambiguity with respect to the angles, and there are two DOAs for 200000 H_Z and three DOAs each for 400000 and 800000 H_Z . For example, there are three DOAs 69.7°, 61.9° and 50.0° corresponding to the frequency 400000 H_Z , so additional computations [13] are needed for determining the true frequency and angle pair. In contrast, neither the EVEV nor the ESEV method requires additional pairing computations.

Frequency (Hz)	ESES			EVEV	ESEV
	Angle 1	Angle2	Angle 3	Angle	Angle
200,000	46.0	20.0	_	20.0	20.0
400,000	69.7	61.9	50.0	50.0	50.0
800,000	80.0	76.4	71.3	80.0	80.0

Table 1: DOA estimation of ESES, EVEV and ESEV

Simulation 2. The performances of the proposed algorithms (EVEV and ESEV) are examined. Figs. 1 and 2 display the outcomes of the EVEV method and ESEV method when the angles of the 10 incoming signals are 20°, 40°, and 60° with the same frequency of 200000 H_Z ; 15°, 30° , 40° , and 50° with the same frequency of $400000 H_Z$; and 40° , 60° and 80° with the same frequency of $800000 H_Z$. From Figs. 1 and 2, we find that the EVEV and ESEV methods can not only obtain the frequency and angle estimations correctly without pairing ambiguity but can also resolve K = 10 (K = M) sources. However, the ESES method cannot handle this case.



Figure 1: Frequency and DOA estimation performance of EVEV



Figure 2: Frequency and DOA estimation performance of ESEV

Simulation 3. The performances of ESES, ESEV and EVEV under different SNRs and different numbers of snapshots are studied in this simulation. All the simulation conditions are the same as those in Simulation 1 except for the following descriptions. The number of snapshots for each experiment is set to 200, and the SNR is changed from 0 to 30 dB for the three sources. Figs. 3 and 4 show the simulation results of the frequency and DOA parameters when the SNR changes, respectively. Then, the SNR is fixed to 20 dB, and the number of snapshots is varied from 100 to 800. In this case, the simulation results of the frequency and DOA parameters are illustrated in Figs. 5 and 6, respectively.

From Figs. 3–6, it can be observed that all three methods are effective for joint frequency and angle estimation. In addition, when the SNR or the number of snapshots increases, the RMSEs of both the frequency and angle estimations decrease. Figs. 3 and 5 illustrate that the frequency estimation performance of the EVEV method is better than that of the ESES method. Additionally, it can be seen from Figs. 4 and 6 that the angle estimation performance of the

ESEV method is slightly better than that of the ESES method. In addition, note that the ESEV method has the same frequency estimation result as that of the ESES method and the same angle estimation result as that of the EVEV method. Therefore, from Figs. 4 and 6, it is inferred that the angle estimation performance of the EVEV method is better than that of the ESES method.



Figure 3: The RMSEs of the frequency estimations vs. the SNR



Figure 4: The RMSEs of the DOA estimations vs. the SNR



Figure 5: The RMSEs of the frequency estimations vs. the number of snapshots



Figure 6: The RMSEs of the DOA estimations vs. the number of snapshots

Simulation 4. The computational complexities of these three methods are compared in this subsection. Suppose M = 10, P = 3 and K = 3, as in Simulations 1 and 3. For simplicity of comparison with other methods, the number of search grids r = 20 for the EVEV method. When the number of snapshots is varied from L = 100 to L = 2000, Fig. 7 shows the computational complexities of the three methods. The EVEV algorithm has the largest computational complexity under this condition. In fact, the computational burden of EVEV becomes heavier as r increases. It can also be seen from Fig. 7 that the ESEV algorithm has almost the same computational complexity as that of the ESES method. In summary, these results coincide with the aforementioned theoretical analysis.



Figure 7: Computational complexity of ESES, ESEV and EVEV

Simulation 5. The effectiveness of ESES, ESEV and EVEV in handling a nonuniform linear array is studied. Assume that eight sensors are located at positions [00.81.72.83.44.55.16.5] and normalized by half of the wavelength. The three sources are from 80° , 50° and 20° with different frequencies for different methods. From Fig. 8, it is verified that all three methods can work well with the nonuniform linear array.



Figure 8: Frequency and DOA estimation for a nonuniform linear array

5 Conclusion

This paper has considered the problem of joint frequency and DOA estimation in array signal processing. By applying the Rayleigh–Ritz theorem to the multiple-delay output model, the proposed methods can achieve automatically paired frequency and DOA estimations without an exhaustive two-dimensional search. The investigations, which were demonstrated in theory and through the use of extensive simulations, have shown that the proposed methods have better estimation performances than those of the ESES and ESEV methods, making a good compromise between estimation accuracy and computational complexity compared with EVEV and ESES. Furthermore, the proposed two methods can always resolve no fewer sources than can ESES. In addition, these methods can also be used for other arrays, including but not limited to nonuniform linear arrays. Future research might be expected to extend the applications of these methods to situations in the presence of multiple coherent sources, array gain-phase errors or mutual coupling effects among the examined sensors.

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