# DISTRIBUTED COVERAGE GAMES FOR ENERGY-AWARE MOBILE SENSOR NETWORKS* 

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#### Abstract

Inspired by current challenges in data-intensive and energy-limited sensor networks, we formulate a coverage optimization problem for mobile sensors as a (constrained) repeated multiplayer game. Each sensor tries to optimize its own coverage while minimizing the processing/energy cost. The sensors are subject to the informational restriction that the environmental distribution function is unknown a priori. We present two distributed learning algorithms where each sensor only remembers its own utility values and actions played during the last plays. These algorithms are proven to be convergent in probability to the set of (constrained) Nash equilibria and global optima of a certain coverage performance metric, respectively. Numerical examples are provided to verify the performance of our proposed algorithms.


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1. Introduction. There is a widespread belief that continuous and pervasive monitoring will be possible in the near future with large numbers of networked, mobile, and wireless sensors. Thus, we are witnessing intense research activity that focuses on the design of efficient control mechanisms for these systems. In particular, distributed algorithms would allow sensor networks to react autonomously to unanticipated changes with minimal human supervision.

A substantial body of research on sensor networks has concentrated on simple sensors that can collect scalar data, e.g., temperature, humidity, or pressure data. A main objective is the design of algorithms that can lead to optimal collective sensing through efficient motion control and communication schemes. However, scalar measurements can be insufficient in many situations, e.g., automated surveillance or traffic monitoring. In contrast, data-intensive sensors such as cameras can collect visual data that are rich in information, thus having tremendous potential for monitoring applications, but at the cost of a higher processing overhead or energy cost.

Precisely, this paper aims to solve a coverage optimization problem taking into account part of the sensing/processing trade-off. Coverage optimization problems have mainly been formulated as cooperative problems where each sensor benefits from sensing the environment as a member of a group. However, sensing may also require expenditure, e.g., the energy consumed or the time spent by data processing algorithms in sensor networks. Because of this, we endow each sensor with a utility function that quantifies this trade-off, formulating a coverage problem as a variation of the congestion games in [29].

[^0]Literature review. In broad terms, the problem studied here is related to a bevy of sensor location and planning problems in the computational geometry, geometric optimization, and robotics literature. Regarding camera networks (a class of sensors that motivated this work), we refer the reader to different variations on the art gallery problem, such as those included in references [27], [31], [35]. The objective here is to find the optimum number of guards in a nonconvex environment so that each point is visible from at least one guard. A related set of references for the deployment of mobile robots with omnidirectional cameras includes [13], [12]. Unlike the art gallery classic algorithms, the latter papers assume that robots have local knowledge of the environment and no recollection of the past. Other related references on robot deployment in convex environments include [8], [18] for anisotropic and circular footprints.

The paper [1] is an excellent survey on multimedia sensor networks where the state of the art in algorithms, protocols, and hardware is shown, and open research issues are discussed in detail. As observed in [9], multimedia sensor networks enhance traditional surveillance systems by enlarging, enhancing, and enabling multiresolution views. The investigation of coverage problems for static visual sensor networks is conducted in [7], [15], [36].

Another set of references relevant to this paper comprise those on the use of game-theoretic tools to (i) solve static target assignment problems, and (ii) devise efficient and secure algorithms for communication networks. In [19], the authors present a game-theoretic analysis of a coverage optimization problem for static sensor networks. This problem is equivalent to the weapon-target assignment problem in [26] which is NP complete. In general, the solution to assignment problems is hard from a combinatorial optimization viewpoint.

Game theory and learning in games are used to analyze a variety of fundamental problems in, e.g., wireless communication networks and the Internet. An incomplete list of references includes [2] on power control, [30] on routing, and [33] on flow control. However, there has been limited research on how to employ learning in games to develop distributed algorithms for mobile sensor networks. One exception is the paper [20], where the authors establish a link between cooperative control problems (in particular, consensus problems) and games (in particular, potential games and weakly acyclic games).

Statement of contributions. The contributions of this paper pertain to both coverage optimization problems and learning in games. Compared with [17] and [18], this paper employs a more accurate sensing model, and the results can be easily extended to include nonconvex environments. Contrary to [17], we do not consider energy expenditure from sensor motions. Furthermore, the algorithms developed allow mobile sensors to self-deploy in the environments where informational distribution functions are unknown a priori.

Regarding learning in games, we extend the use of the payoff-based learning dynamics first novelly proposed in [21], [22]. The coverage game we consider here is shown to be a (constrained) exact potential game. A number of learning rules, e.g., better (or best) reply dynamics and adaptive play, have been proposed to reach Nash equilibria in potential games. In these algorithms, each player must have access to the utility values induced by alternative actions. In our problem setup, however, this information is inaccessible because of the information constraints caused by unknown rewards, motion, and sensing limitations. To tackle this challenge, we develop two distributed payoff-based learning algorithms where each sensor remembers only its own utility values and actions played during the last two plays.

In the first algorithm, at each time step, each sensor repeatedly updates its action synchronously, either trying some new action in the state-dependent feasible action set or selecting the action which corresponds to a higher utility value in the most recent two time steps. Like the algorithm for the special identical interest games in [22], the first algorithm employs a diminishing exploration rate. The dynamically changing exploration rate renders the algorithm a time-inhomogeneous Markov chain and allows for the convergence in probability to the set of (constrained) Nash equilibria, from which no agent is willing to unilaterally deviate.

The second algorithm is asynchronous. At each time step, only one sensor is active and updates its state by either trying some new action in the state-dependent feasible action set or selecting an action according to a Gibbs-like distribution from those played in the last two time steps when it was active. The algorithm is shown to be convergent in probability to the set of global maxima of a coverage performance metric. Rather than maximizing the associated potential function in [21], the second algorithm optimizes the sum of local utility functions, which better capture a global trade-off between the overall network benefit from sensing and the total energy the network consumes. By employing a diminishing exploration rate, our algorithm is guaranteed to have stronger convergence properties than the ones in [21]. A more detailed comparison with [21], [22] is provided in Remarks 3.2 and 3.4. The results in the current paper were presented in [39], [40], where the analysis of numerical examples and the technical details were omitted or summarized.
2. Problem formulation. Here, we first review some basic game-theoretic concepts; see, for example, [11]. This will allow us to formulate subsequently an optimal coverage problem for mobile sensor networks as a repeated multiplayer game. We then introduce notation used throughout the paper.
2.1. Background in game theory. A strategic game $\Gamma:=\langle V, A, U\rangle$ has three components:

1. A set $V$ enumerating players $i \in V:=\{1, \ldots, N\}$.
2. An action set $A:=\prod_{i=1}^{N} A_{i}$, the space of all actions vectors, where $s_{i} \in A_{i}$ is the action of player $i$ and a (multiplayer) action $s \in A$ has components $s_{1}, \ldots, s_{N}$.
3. The collection of utility functions $U$, where the utility function $u_{i}: A \rightarrow \mathbb{R}$ models player $i$ 's preferences over action profiles.
Denote by $s_{-i}$ the action profile of all players other than $i$, and by $A_{-i}=\prod_{j \neq i} A_{j}$ the set of action profiles for all players except $i$. The concept of (pure) Nash equilibrium (NE, for short) is the most important one in noncooperative game theory [11] and is defined as follows.

Definition 2.1 (Nash equilibrium [11]). Consider the strategic game $\Gamma$. An action profile $s^{*}:=\left(s_{i}^{*}, s_{-i}^{*}\right)$ is a (pure) NE of the game $\Gamma$ if for all $i \in V$ and for all $s_{i} \in A_{i}$ it holds that $u_{i}\left(s^{*}\right) \geq u_{i}\left(s_{i}, s_{-i}^{*}\right)$.

An action profile corresponding to an NE represents a scenario where no player has incentive to unilaterally deviate. Exact potential games form an important class of strategic games where the change in a player's utility caused by a unilateral deviation can be exactly measured by a potential function.

Definition 2.2 (exact potential game [25]). The strategic game $\Gamma$ is an exact potential game with potential function $\phi: A \rightarrow \mathbb{R}$ if, for every $i \in V$, for every $s_{-i} \in A_{-i}$, and for every $s_{i}, s_{i}^{\prime} \in A_{i}$, it holds that

$$
\begin{equation*}
\phi\left(s_{i}, s_{-i}\right)-\phi\left(s_{i}^{\prime}, s_{-i}\right)=u_{i}\left(s_{i}, s_{-i}\right)-u_{i}\left(s_{i}^{\prime}, s_{-i}\right) \tag{2.1}
\end{equation*}
$$

In conventional noncooperative game theory, all the actions in $A_{i}$ can always be selected by player $i$ in response to other players' actions. However, in the context of motion coordination, the actions available to player $i$ will often be constrained to a state-dependent subset of $A_{i}$. In particular, we denote by $F_{i}\left(s_{i}, s_{-i}\right) \subseteq A_{i}$ the set of feasible actions of player $i$ when the action profile is $s:=\left(s_{i}, s_{-i}\right)$. We assume that $s_{i} \in F_{i}\left(s_{i}, s_{-i}\right)$ for any $s \in A$ throughout this paper. Denote $F(s):=\prod_{i \in V} F_{i}(s) \subseteq A$ for all $s \in A$ and $F:=\cup\{F(s) \mid s \in A\}$. The introduction of $F$ naturally leads to the notion of constrained strategic game $\Gamma_{\text {res }}:=\langle V, A, U, F\rangle$ and the following associated concepts.

Definition 2.3 (constrained Nash equilibrium). Consider the constrained strategic game $\Gamma_{\text {res }}$. An action profile s* is a constrained (pure) NE of the game $\Gamma_{\text {res }}$ if for all $i \in V$ and for all $s_{i} \in F_{i}\left(s_{i}^{*}, s_{-i}^{*}\right)$ it holds that $u_{i}\left(s^{*}\right) \geq u_{i}\left(s_{i}, s_{-i}^{*}\right)$.

Definition 2.4 (constrained exact potential game). The game $\Gamma_{\text {res }}$ is a constrained exact potential game with potential function $\phi(s)$ if, for every $i \in V$, every $s_{-i} \in A_{-i}$, and every $s_{i} \in A_{i}$, the equality (2.1) holds for every $s_{i}^{\prime} \in F_{i}\left(s_{i}, s_{-i}\right)$.

With the assumption of $s_{i} \in F_{i}\left(s_{i}, s_{-i}\right)$ for any $s \in A$, we observe that if $s^{*}$ is an NE of the strategic game $\Gamma$, then it is also a constrained NE of the constrained strategic game $\Gamma_{\text {res }}$. For any given strategic game, NE may not exist. However, the existence of NE in exact potential games is guaranteed [25]. Hence, any constrained exact potential game with the assumption of $s_{i} \in F_{i}\left(s_{i}, s_{-i}\right)$ for any $s \in A$ has at least one constrained NE.

### 2.2. Coverage problem formulation.

2.2.1. Mission space. We consider a convex two-dimensional mission space that is discretized into a (squared) lattice. We assume that each square of the lattice has unit dimensions. Each square will be labeled with the coordinate of its center $q=$ $\left(q_{x}, q_{y}\right)$, where $q_{x} \in\left[q_{x_{\min }}, q_{x_{\text {max }}}\right]$ and $q_{y} \in\left[q_{y_{\text {min }}}, q_{y_{\text {max }}}\right]$, for some integers $q_{x_{\text {min }}}, q_{y_{\text {min }}}$, $q_{x_{\max }}, q_{y_{\max }}$. Denote by $\mathcal{Q}$ the collection of all squares of the lattice.

We now define an associated location graph $\mathcal{G}_{\text {loc }}:=\left(\mathcal{Q}, E_{\text {loc }}\right)$ where $\left(\left(q_{x}, q_{y}\right)\right.$, $\left.\left(q_{x^{\prime}}, q_{y^{\prime}}\right)\right) \in E_{\text {loc }}$ if and only if $\left|q_{x}-q_{x^{\prime}}\right|+\left|q_{y}-q_{y^{\prime}}\right|=1$ for $\left(q_{x}, q_{y}\right),\left(q_{x^{\prime}}, q_{y^{\prime}}\right) \in \mathcal{Q}$. Note that the graph $\mathcal{G}_{\text {loc }}$ is undirected; i.e., $\left(q, q^{\prime}\right) \in E_{\text {loc }}$ if and only if $\left(q^{\prime}, q\right) \in E_{\text {loc }}$. The set of neighbors of $q$ in $E_{\text {loc }}$ is given by $\mathcal{N}_{q}^{\text {loc }}:=\left\{q^{\prime} \in \mathcal{Q} \backslash\{q\} \mid\left(q, q^{\prime}\right) \in E_{\text {loc }}\right\}$. We assume that $\mathcal{G}_{\text {loc }}$ is fixed and connected, and we denote its diameter by $D$.

Agents are deployed in $\mathcal{Q}$ to detect certain events of interest. As agents move in $\mathcal{Q}$ and process measurements, they will assign a numerical value $W_{q} \geq 0$ to the events in each square with center $q \in \mathcal{Q}$. If $W_{q}=0$, then there is no significant event at the square with center $q$. The larger the value of $W_{q}$ is, the more pertinent the set of events at the square with center $q$ is. Later, the amount $W_{q}$ will be identified with a benefit of observing the point $q$. In this setup, we assume the values $W_{q}$ to be constant in time. Furthermore, $W_{q}$ is not a prior knowledge to the agents, but the agents can measure this value through sensing the point $q$.
2.2.2. Modeling of the energy-limited sensor nodes. Each mobile agent $i$ is modeled as a point mass in $\mathcal{Q}$, with location $a_{i}:=\left(x_{i}, y_{i}\right) \in \mathcal{Q}$. Each agent has mounted a sensor on board, whose operation has an energy/processing cost roughly proportional to the limited area that it scans. Examples include line-of-sight sensors such as radar and sonar-like sensors where energy spent is proportional to the monitored area that is scanned with the beam. Another example that motivated this work was that of cameras, with a limited visual footprint and processing cost proportional to the camera footprint.


Fig. 2.1. Sensor footprint and a configuration of the mobile sensor network.

We assume that the footprint of the sensor is directional, has limited range, and has a finite "angle of view" centered at the sensor position. Following a geometric simplification, we model the sensing region of agent $i$ as an annulus sector in the two-dimensional plane; see Figure 2.1.

The sensor footprint is completely characterized by the following parameters: the position of agent $i, a_{i} \in \mathcal{Q}$; the sensor orientation, given by an angle $\theta_{i} \in[0,2 \pi)$; the angle of view, $\zeta_{i} \in\left[\zeta_{\min }, \zeta_{\max }\right]$; and the shortest range (resp., longest range) between agent $i$ and the nearest (resp., farthest) object that can be identified by scanning the sensor footprint, $r_{i}^{\text {shrt }} \in\left[r_{\min }, r_{\max }\right]$ (resp., $r_{i}^{\operatorname{lng}} \in\left[r_{\min }, r_{\max }\right]$ ). We assume the parameters $r_{i}^{\text {shrt }}, r_{i}^{\operatorname{lng}}, \zeta_{i}$ are sensor parameters that can be chosen. In this way, $c_{i}:=\left(\mathrm{FL}_{i}, \theta_{i}\right) \in\left[0, \mathrm{FL}_{\text {max }}\right] \times[0,2 \pi)$ is the sensor control vector of agent $i$. In what follows, we will assume that $c_{i}$ takes values in a finite subset $\mathcal{C} \subset\left[0, \mathrm{FL}_{\max }\right] \times[0,2 \pi)$. An agent action is thus a vector $s_{i}:=\left(a_{i}, c_{i}\right) \in \mathcal{A}_{i}:=\mathcal{Q} \times \mathcal{C}$, and a multiagent action is denoted by $s=\left(s_{1}, \ldots, s_{N}\right) \in \mathcal{A}:=\prod_{i=1}^{N} \mathcal{A}_{i}$.

Let $\mathcal{D}\left(a_{i}, c_{i}\right)$ be the sensor footprint of agent $i$. Now we can define a proximity sensing graph ${ }^{1} \mathcal{G}_{\text {sen }}(s):=\left(V, E_{\text {sen }}(s)\right)$ as follows: the set of neighbors of agent $i$, $\mathcal{N}_{i}^{\text {sen }}(s)$, is given as $\mathcal{N}_{i}^{\text {sen }}(s):=\left\{j \in V \backslash\{i\} \mid \mathcal{D}\left(a_{i}, c_{i}\right) \cap \mathcal{D}\left(a_{j}, c_{j}\right) \cap \mathcal{Q} \neq \emptyset\right\}$.

Each agent is able to communicate with others to exchange information. We assume that the communication range of agents is $2 r_{\max }$. This induces a $2 r_{\max }$-disk communication graph $\mathcal{G}_{\text {comm }}(s):=\left(V, E_{\text {comm }}(s)\right)$ as follows: the set of neighbors of agent $i$ is given by $\mathcal{N}_{i}^{\text {comm }}(s):=\left\{j \in V \backslash\{i\} \mid\left(x_{i}-x_{j}\right)^{2}+\left(y_{i}-y_{j}\right)^{2} \leq\left(2 r_{\text {max }}\right)^{2}\right\}$. Note that $\mathcal{G}_{\text {comm }}(s)$ is undirected and that $\mathcal{G}_{\text {sen }}(s) \subseteq \mathcal{G}_{\text {comm }}(s)$.

The motion of agents will be limited to a neighboring point in $\mathcal{G}_{\text {loc }}$ at each time step. Thus, an agent feasible action set will be given by $\mathcal{F}_{i}\left(a_{i}\right):=\left(\left\{a_{i}\right\} \cup \mathcal{N}_{a_{i}}^{\text {loc }}\right) \times \mathcal{C}$.
2.2.3. Coverage game. We now proceed to formulate a coverage optimization problem as a constrained strategic game. For each $q \in \mathcal{Q}$, we denote $n_{q}(s)$ as the cardinality of the set $\left\{k \in V \mid q \in \mathcal{D}\left(a_{k}, c_{k}\right) \cap \mathcal{Q}\right\}$, i.e., the number of agents which can observe the point $q$. The "profit" given by $W_{q}$ will be equally shared by agents that can observe the point $q$. The benefit that agent $i$ obtains through sensing is thus defined by $\sum_{q \in \mathcal{D}\left(a_{i}, c_{i}\right) \cap \mathcal{Q}} \frac{W_{q}}{n_{q}(s)}$.

[^1]In the following, we associate an energy/processing cost with the use of the sensor: $f_{i}\left(c_{i}\right):=\frac{1}{2} \zeta_{i}\left(\left(r_{i}^{\text {lng }}\right)^{2}-\left(r_{i}^{\text {shrt }}\right)^{2}\right)$.

As mentioned earlier, this can be representative of radar, sonar-like sensors, laserrange finders, and camera-like sensors. For example, while recent efforts have been dedicated to designing radar with an adjustable field of view or beam [37], typical search radar systems (similarly, sonars) are swept over the areas to be scanned with rotating antennas [6]. Energy spent is proportional to the scanned sectors which are controlled through the rotation of the antenna. Camera sensors which, on the other hand, are passive sensors require a processing cost that is also proportional to the area to be scanned [23]. Motivated by this, [34] proposes to save energy in videobased sensor networks by partitioning the sensing task among sensors with overlapping fields of view. This effectively translates into a "reduction" of the area to be scanned by cameras or reduction in their field of view. In the paper [28] another coverage algorithm is proposed to turn off some sensors with overlapping fields of view. We also remark that the algorithms and analysis that follow later are not affected by whether or not the energy cost function is removed. In fact, several papers in the visual sensor network literature are devoted to maximizing the coverage provided by the sum of the areas of the sensor footprints; see [15], [36] and the references therein. Our algorithms provide solutions to those coverage problems when sensors are mobile and the informational distribution function is unknown a priori.

We will endow each agent with a utility function that aims to capture the above sensing/processing trade-off. In this way, we define a utility function for agent $i$ by

$$
u_{i}(s)=\sum_{q \in \mathcal{D}\left(a_{i}, c_{i}\right) \cap \mathcal{Q}} \frac{W_{q}}{n_{q}(s)}-f_{i}\left(c_{i}\right)
$$

Note that the utility function $u_{i}$ is local over the sensing graph $\mathcal{G}_{\text {sen }}(s)$; i.e., $u_{i}$ is only dependent on the actions of $\{i\} \cup \mathcal{N}_{i}^{\text {sen }}(s)$. With the set of utility functions $U_{\text {cov }}=\left\{u_{i}\right\}_{i \in V}$ and feasible action set $\mathcal{F}_{\text {cov }}=\Pi_{i=1}^{N} \bigcup_{a_{i} \in \mathcal{A}_{i}} \mathcal{F}_{i}\left(a_{i}\right)$, we now have all the ingredients to introduce the coverage game $\Gamma_{\text {cov }}:=\left\langle V, \mathcal{A}, U_{\text {cov }}, \mathcal{F}_{\text {cov }}\right\rangle$. This game is a variation of the congestion games introduced in [29].

LEMMA 2.5. The coverage game $\Gamma_{\text {cov }}$ is a constrained exact potential game with potential function

$$
\phi(s)=\sum_{q \in \mathcal{Q}} \sum_{\ell=1}^{n_{q}(s)} \frac{W_{q}}{\ell}-\sum_{i=1}^{N} f_{i}\left(c_{i}\right)
$$

Proof. The proof is a slight variation of that in [29]. Consider any $s:=\left(s_{i}, s_{-i}\right) \in$ $\mathcal{A}$, where $s_{i}:=\left(a_{i}, c_{i}\right)$. We fix $i \in V$ and pick any $s_{i}^{\prime}=\left(a_{i}^{\prime}, c_{i}^{\prime}\right)$ from $\mathcal{F}_{i}\left(a_{i}\right)$. Denote $s^{\prime}:=\left(s_{i}^{\prime}, s_{-i}\right), \Omega_{1}:=\left(\mathcal{D}\left(a_{i}, c_{i}\right) \backslash \mathcal{D}\left(a_{i}^{\prime}, c_{i}^{\prime}\right)\right) \cap \mathcal{Q}$, and $\Omega_{2}:=\left(\mathcal{D}\left(a_{i}^{\prime}, c_{i}^{\prime}\right) \backslash \mathcal{D}\left(a_{i}, c_{i}\right)\right) \cap \mathcal{Q}$. Observe that

$$
\begin{aligned}
& \phi\left(s_{i}, s_{-i}\right)-\phi\left(s_{i}^{\prime}, s_{-i}\right) \\
& =\sum_{q \in \Omega_{1}}\left(\sum_{\ell=1}^{n_{q}(s)} \frac{W_{q}}{\ell}-\sum_{\ell=1}^{n_{q}\left(s^{\prime}\right)} \frac{W_{q}}{\ell}\right)+\sum_{q \in \Omega_{2}}\left(-\sum_{\ell=1}^{n_{q}(s)} \frac{W_{q}}{\ell}+\sum_{\ell=1}^{n_{q}\left(s^{\prime}\right)} \frac{W_{q}}{\ell}\right)-f_{i}\left(c_{i}\right)+f_{i}\left(c_{i}^{\prime}\right) \\
& =\sum_{q \in \Omega_{1}} \frac{W_{q}}{n_{q}(s)}-\sum_{q \in \Omega_{2}} \frac{W_{q}}{n_{q}\left(s^{\prime}\right)}-f_{i}\left(c_{i}\right)+f_{i}\left(c_{i}^{\prime}\right) \\
& =u_{i}\left(s_{i}, s_{-i}\right)-u_{i}\left(s_{i}^{\prime}, s_{-i}\right)
\end{aligned}
$$

where in the second equality we utilize the fact that for each $q \in \Omega_{1}, n_{q}(s)=n_{q}\left(s^{\prime}\right)+1$, and for each $q \in \Omega_{2}, n_{q}\left(s^{\prime}\right)=n_{q}(s)+1$.

We denote by $\mathcal{E}\left(\Gamma_{\text {cov }}\right)$ the set of constrained NEs of $\Gamma_{\text {cov }}$. It is worth mentioning that $\mathcal{E}\left(\Gamma_{\text {cov }}\right) \neq \emptyset$ due to the fact that $\Gamma_{\text {cov }}$ is a constrained exact potential game.

Remark 2.1. The assumptions of our problem formulation admit several extensions. For example, it is straightforward to extend our results to nonconvex threedimensional spaces. This is because the results that follow can also handle other shapes of the sensor footprint, e.g., a complete disk, a subset of the annulus sector. On the other hand, note that the coverage problem can be interpreted as a target assignment problem-here, the value $W_{q} \geq 0$ would be associated with the value of a target located at the point $q$.
2.3. Notations. In the following, we will use the Landau symbol, $O$, as in $O\left(\epsilon^{k}\right)$ for some $k \geq 0$. This implies that $0<\lim _{\epsilon \rightarrow 0^{+}} \frac{O\left(\epsilon^{k}\right)}{\epsilon^{k}}<+\infty$. We denote $\operatorname{diag} \mathcal{A}:=\left\{(s, s) \in \mathcal{A}^{2} \mid s \in \mathcal{A}\right\}$ and $\operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right):=\left\{(s, s) \in \mathcal{A}^{2} \mid s \in \mathcal{E}\left(\Gamma_{\text {cov }}\right)\right\}$.

Consider $a, a^{\prime} \in \mathcal{Q}^{N}$, where $a_{i} \neq a_{i}^{\prime}$ and $a_{-i}=a_{-i}^{\prime}$ for some $i \in V$. The transition $a \rightarrow a^{\prime}$ is feasible if and only if $\left(a_{i}, a_{i}^{\prime}\right) \in E_{\text {loc }}$. A feasible path from $a$ to $a^{\prime}$ consisting of multiple feasible transitions is denoted by $a \Rightarrow a^{\prime}$. Let $\diamond a:=\left\{a^{\prime} \in \mathcal{Q} \mid a \Rightarrow a^{\prime}\right\}$ be the reachable set from $a$.

Let $s=(a, c), s^{\prime}=\left(a^{\prime}, c^{\prime}\right) \in \mathcal{A}$, where $a_{i} \neq a_{i}^{\prime}$ and $a_{-i}=a_{-i}^{\prime}$ for some $i \in V$. The transition $s \rightarrow s^{\prime}$ is feasible if and only if $s_{i}^{\prime} \in \mathcal{F}_{i}(a)$. A feasible path from $s$ to $s^{\prime}$ consisting of multiple feasible transitions is denoted by $s \Rightarrow s^{\prime}$. Finally, $\diamond s:=\left\{s^{\prime} \in \mathcal{A} \mid s \Rightarrow s^{\prime}\right\}$ will be the reachable set from $s$.
3. Distributed coverage learning algorithms and convergence results. In our coverage problem, we assume that $W_{q}$ is unknown to all the sensors in advance. Furthermore, due to the restrictions of motion and sensing, each agent is unable to obtain the information of $W_{q}$ if the point $q$ is outside its sensing range. In addition, the utility of each agent depends upon the group strategy. These information constraints render each agent unable to access the utility values induced by alternative actions. Thus the action-based learning algorithms, e.g., better (or best) reply learning algorithm and adaptive play learning algorithm, cannot be employed to solve our coverage games. It motivates us to design distributed learning algorithms which only require the payoff received.

In this section, we come up with two distributed payoff-based learning algorithms, the distributed inhomogeneous synchronous coverage learning algorithm (DISCL, for short) and distributed inhomogeneous asynchronous coverage learning algorithm (DIACL, for short). We then present their convergence properties. Relevant algorithms include payoff-based learning algorithms proposed in [21], [22].
3.1. Distributed inhomogeneous synchronous coverage learning algorithm. For each $t \geq 1$ and $i \in V$, we define $\tau_{i}(t)$ as follows: $\tau_{i}(t)=t$ if $u_{i}(s(t)) \geq$ $u_{i}(s(t-1))$; otherwise, $\tau_{i}(t)=t-1$. Here, $s_{i}\left(\tau_{i}(t)\right)$ is the more successful action of agent $i$ in the last two steps. The DISCL algorithm is formally stated in the following:
1: [Initialization:] At $t=0$, all agents are uniformly placed in $\mathcal{Q}$. Each agent $i$ uniformly chooses its camera control vector $c_{i}$ from the set $\mathcal{C}$, communicates with agents in $\mathcal{N}_{i}^{\text {sen }}(s(0))$, and computes $u_{i}(s(0))$. At $t=1$, all the agents keep their actions.
2: [Update:] At each time $t \geq 2$, each agent $i$ updates its state according to the following rules:

- Agent $i$ chooses the exploration rate $\epsilon(t)=t^{-\frac{1}{N(D+1)}}$, with $D$ being the diameter of the location graph $\mathcal{G}_{\text {loc }}$, and computes $s_{i}\left(\tau_{i}(t)\right)$.
- With probability $\epsilon(t)$, agent $i$ experiments and chooses the temporary action $s_{i}^{\mathrm{tp}}:=\left(a_{i}^{\mathrm{tp}}, c_{i}^{\mathrm{tp}}\right)$ uniformly from the set $\mathcal{F}_{i}\left(a_{i}(t)\right) \backslash\left\{s_{i}\left(\tau_{i}(t)\right)\right\}$.
- With probability $1-\epsilon(t)$, agent $i$ does not experiment and sets $s_{i}^{\text {tp }}=$ $s_{i}\left(\tau_{i}(t)\right)$.
- After $s_{i}^{\mathrm{tp}}$ is chosen, agent $i$ moves to the position $a_{i}^{\mathrm{tp}}$ and sets the camera control vector to $c_{i}^{\mathrm{tp}}$.
3: [Communication and computation:] At position $a_{i}^{\text {tp }}$, each agent $i$ sends the information $\mathcal{D}\left(a_{i}^{\mathrm{tp}}, c_{i}^{\mathrm{tp}}\right) \cap \mathcal{Q}$ to agents in $\mathcal{N}_{i}^{\text {sen }}\left(s_{i}^{\mathrm{tp}}, s_{-i}^{\mathrm{tp}}\right)$. After that, each agent $i$ identifies the quantity $n_{q}\left(s^{\mathrm{tp}}\right)$ for each $q \in \mathcal{D}\left(a_{i}^{\mathrm{tp}}, c_{i}^{\mathrm{tp}}\right) \cap \mathcal{Q}$ and computes the utility $u_{i}\left(s_{i}^{\mathrm{tp}}, s_{-i}^{\mathrm{tp}}\right)$ and the feasible action set of $\mathcal{F}_{i}\left(a_{i}^{\mathrm{tp}}\right)$.
4: Repeat steps 2 and 3.
Remark 3.1. A variation of the DISCL algorithm corresponds to $\epsilon(t)=\epsilon \in\left(0, \frac{1}{2}\right]$ constant for all $t \geq 2$. If this is the case, we will refer to the algorithm as distributed homogeneous synchronous coverage learning algorithm (DHSCL, for short). Later, the convergence analysis of the DISCL algorithm will be based on the analysis of the DHSCL algorithm.

Denote the space $\mathcal{B}:=\left\{\left(s, s^{\prime}\right) \in \mathcal{A} \times \mathcal{A} \mid s_{i}^{\prime} \in \mathcal{F}_{i}\left(a_{i}\right) \forall i \in V\right\}$. Observe that $z(t):=(s(t-1), s(t))$ in the DISCL algorithm constitutes a time-inhomogeneous Markov chain $\left\{\mathcal{P}_{t}\right\}$ on the space $\mathcal{B}$. The following theorem states that the DISCL algorithm asymptotically converges to the set of $\mathcal{E}\left(\Gamma_{\text {cov }}\right)$ in probability.

Theorem 3.1. Consider the Markov chain $\left\{\mathcal{P}_{t}\right\}$ induced by the DISCL algorithm. It holds that $\lim _{t \rightarrow+\infty} \mathbb{P}\left(z(t) \in \operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)\right)=1$.

The proofs of Theorem 3.1 are provided in section 4.
Remark 3.2. An algorithm is proposed for the general class of weakly acyclic games (including potential games as special cases) in [22] and is able to find an NE with an arbitrarily high probability by choosing an arbitrarily small and fixed exploration rate $\epsilon$ in advance. However, it is difficult to derive an analytic relation between the convergent probability and the exploration rate. For the special case of identical interest games (all players share an identical utility function), the authors in [22] exploit a diminishing exploration rate and obtain a stronger result of convergence in probability. This motivates us to utilize a diminishing exploration rate in the DISCL algorithm which allows for the convergence to the set of NEs in probability. In the algorithm for weakly acyclic games in [22], each player may execute the baseline action which depends on all the past plays. As a result, the algorithm for weakly acyclic games in [22] cannot be utilized to solve our problem because the baseline action may not be feasible when the state-dependent constraints are present. It is worth mentioning that [22] studies a case where the utilities are corrupted by noises.
3.2. Distributed inhomogeneous asynchronous coverage learning algorithm. Lemma 2.5 shows that the coverage game $\Gamma_{\text {cov }}$ is a constrained exact potential game with potential function $\phi(s)$. However, this potential function is not a straightforward measure of the network coverage performance. On the other hand, the objective function $U_{g}(s):=\sum_{i \in V} u_{i}(s)$ captures the trade-off between the overall network benefit from sensing and the total energy the network consumes, and thus can be perceived as a more natural coverage performance metric. Denote $S^{*}:=\left\{s \mid \operatorname{argmax}_{s \in \mathcal{A}} U_{g}(s)\right\}$ as the set of global maximizers of $U_{g}(s)$. In this part, we present the DIACL algorithm, which is convergent in probability to the set $S^{*}$.

Before doing this, we first introduce some notation for the DIACL algorithm. Denote by $\mathcal{B}^{\prime}$ the space $\mathcal{B}^{\prime}:=\left\{\left(s, s^{\prime}\right) \in \mathcal{A} \times \mathcal{A} \mid s_{-i}=s_{-i}^{\prime}, s_{i}^{\prime} \in \mathcal{F}_{i}\left(a_{i}\right)\right.$ for some $i \in$ $V\}$. For any $s^{0}, s^{1} \in \mathcal{A}$ with $s_{-i}^{0}=s_{-i}^{1}$ for some $i \in V$, we denote

$$
\Delta_{i}\left(s^{1}, s^{0}\right):=\frac{1}{2} \sum_{q \in \Omega_{1}} \frac{W_{q}}{n_{q}\left(s^{1}\right)}-\frac{1}{2} \sum_{q \in \Omega_{2}} \frac{W_{q}}{n_{q}\left(s^{0}\right)}
$$

where $\Omega_{1}:=\mathcal{D}\left(a_{i}^{1}, c_{i}^{1}\right) \backslash \mathcal{D}\left(a_{i}^{0}, c_{i}^{0}\right) \cap \mathcal{Q}$ and $\Omega_{2}:=\mathcal{D}\left(a_{i}^{0}, c_{i}^{0}\right) \backslash \mathcal{D}\left(a_{i}^{1}, c_{i}^{1}\right) \cap \mathcal{Q}$, and

$$
\begin{aligned}
& \rho_{i}\left(s^{0}, s^{1}\right):=u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)-u_{i}\left(s^{0}\right)+\Delta_{i}\left(s^{0}, s^{1}\right), \\
& \Psi_{i}\left(s^{0}, s^{1}\right):=\max \left\{u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right), u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right\}, \\
& m^{*}:=\max _{\left(s^{0}, s^{1}\right) \in \mathcal{B}, s_{i}^{0} \neq s_{i}^{1}}\left\{\Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right)\right), \frac{1}{2}\right\} .
\end{aligned}
$$

It is easy to check that $\Delta_{i}\left(s^{1}, s^{0}\right)=-\Delta_{i}\left(s^{0}, s^{1}\right)$ and $\Psi_{i}\left(s^{0}, s^{1}\right)=\Psi_{i}\left(s^{1}, s^{0}\right)$. Assume that at each time instant, one of the agents becomes active with equal probability. This can be realized by employing the asynchronous time model proposed in [4] where each node has a clock which ticks according to a rate 1 Poisson process. For this reason, we will refer to the following algorithm as being asynchronous. Denote by $\gamma_{i}(t)$ the last time instant before $t$ when agent $i$ was active. We then denote $\gamma_{i}^{(2)}(t):=\gamma_{i}\left(\gamma_{i}(t)\right)$. The main steps of the DIACL algorithm are described in the following:
1: [Initialization:] At $t=0$, all agents are uniformly placed in $\mathcal{Q}$. Each agent $i$ uniformly chooses the camera control vector $c_{i}$ from the set $\mathcal{C}$ and then communicates with agents in $\mathcal{N}_{i}^{\text {sen }}(s(0))$ and computes $u_{i}(s(0))$. Furthermore, each agent $i$ chooses $m_{i} \in\left(2 m^{*}, K m^{*}\right]$ for some $K \geq 2$. At $t=1$, all the sensors keep their actions.
2: [Update:] Assume that agent $i$ is active at time $t \geq 2$. Then agent $i$ updates its state according to the following rules:

- Agent $i$ chooses the exploration rate $\epsilon(t)=t^{-\frac{1}{(D+1)(K+1) m^{*}}}$.
- With probability $\epsilon(t)^{m_{i}}$, agent $i$ experiments and uniformly chooses $s_{i}^{\mathrm{tp}}:=$ $\left(a_{i}^{\mathrm{tp}}, c_{i}^{\mathrm{tp}}\right)$ from the action set $\mathcal{F}_{i}\left(a_{i}(t)\right) \backslash\left\{s_{i}(t), s_{i}\left(\gamma_{i}^{(2)}(t)+1\right)\right\}$.
- With probability $1-\epsilon(t)^{m_{i}}$, agent $i$ does not experiment and chooses $s_{i}^{\text {tp }}$ according to the following probability distribution:

$$
\begin{aligned}
& \mathbb{P}\left(s_{i}^{\mathrm{tp}}=s_{i}(t)\right)=\frac{1}{1+\epsilon(t)^{\rho_{i}\left(s_{i}\left(\gamma_{i}^{(2)}(t)+1\right), s_{i}(t)\right)}} \\
& \mathbb{P}\left(s_{i}^{\mathrm{tp}}=s_{i}\left(\gamma_{i}^{(2)}(t)+1\right)\right)=\frac{\epsilon(t)^{\rho_{i}\left(s_{i}\left(\gamma_{i}^{(2)}(t)+1\right), s_{i}(t)\right)}}{1+\epsilon(t)^{\rho_{i}\left(s_{i}\left(\gamma_{i}^{(2)}(t)+1\right), s_{i}(t)\right)}}
\end{aligned}
$$

- After $s_{i}^{\mathrm{tp}}$ is chosen, agent $i$ moves to the position $a_{i}^{\mathrm{tp}}$ and sets its camera control vector to be $c_{i}^{\mathrm{tp}}$.
3: [Communication and computation:] At position $a_{i}^{\text {tp }}$, the active agent $i$ initiates a message to agents in $\mathcal{N}_{i}^{\operatorname{sen}}\left(s_{i}^{\mathrm{tp}}, s_{-i}(t)\right)$. Then each agent $j \in \mathcal{N}_{i}^{\mathrm{sen}}\left(s_{i}^{\mathrm{tp}}, s_{-i}(t)\right)$ sends the information of $\mathcal{D}\left(a_{j}^{\mathrm{tp}}, c_{j}^{\mathrm{tp}}\right) \cap \mathcal{Q}$ to agent $i$. After receiving such information, agent $i$ identifies the quantity $n_{q}\left(s_{i}^{\mathrm{tp}}, s_{-i}(t)\right)$ for each $q \in \mathcal{D}\left(a_{i}^{\mathrm{tp}}, c_{i}^{\mathrm{tp}}\right) \cap \mathcal{Q}$ and computes the utility $u_{i}\left(s_{i}^{\mathrm{tp}}, s_{-i}(t)\right), \Delta_{i}\left(\left(s_{i}^{\mathrm{tp}}, s_{-i}(t)\right), s\left(\gamma_{i}(t)+1\right)\right)$, and the feasible action set of $\mathcal{F}_{i}\left(a_{i}^{\mathrm{tp}}\right)$.
4: Repeat steps 2 and 3.

Remark 3.3. A variation of the DIACL algorithm corresponds to $\epsilon(t)=\epsilon \in\left(0, \frac{1}{2}\right]$ constant for all $t \geq 2$. If this is the case, we will refer to the algorithm as the distributed homogeneous asynchronous coverage learning algorithm (DHACL, for short). Later, we will base the convergence analysis of the DIACL algorithm on that of the DHACL algorithm.

Like the DISCL algorithm, $z(t):=(s(t-1), s(t))$ in the DIACL algorithm constitutes a time-inhomogeneous Markov chain $\left\{\mathcal{P}_{t}\right\}$ on the space $\mathcal{B}^{\prime}$. The following theorem states the convergence property of the DIACL algorithm.

Theorem 3.2. Consider the Markov chain $\left\{\mathcal{P}_{t}\right\}$ induced by the DIACL algorithm for the game $\Gamma_{\text {cov }}$. Then it holds that $\lim _{t \rightarrow+\infty} \mathbb{P}\left(z(t) \in \operatorname{diag} S^{*}\right)=1$.

The proof of Theorem 3.2 is provided in section 4.
Remark 3.4. A synchronous payoff-based, log-linear learning algorithm is proposed in [21] for potential games in which players aim to maximize the potential function of the game. As we mentioned before, the potential function is not suitable to act as a coverage performance metric. As opposed to [21], the DIACL algorithm instead seeks to optimize a different function $U_{g}(s)$ perceived as a natural network performance metric. Furthermore, the DIACL algorithm exploits a diminishing stepsize, and this choice allows for convergence to the set of global optima in probability. On the other hand, convergence in [21] is to the set of NE with arbitrarily high probability. Theoretically, our result is stronger than that of [21] by choosing an arbitrarily small and fixed exploration rate in advance.
4. Convergence analysis. In this section, we prove Theorems 3.1 and 3.2 by appealing to the theory of resistance trees in [38] and the results in strong ergodicity in [16]. Relevant papers include [21], [22], where the theory of resistance trees in [38] is novelly utilized to study the class of payoff-based learning algorithms, and [3], [14], [24], where the strong ergodicity theory is employed to characterize the convergence properties of time-inhomogeneous Markov chains.
4.1. Convergence analysis of the DISCL algorithm. We first utilize Theorem 7.6 to characterize the convergence properties of the associated DHSCL algorithm. This is essential for the analysis of the DISCL algorithm.

Observe that $z(t):=(s(t-1), s(t))$ in the DHSCL algorithm constitutes a timehomogeneous Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ on the space $\mathcal{B}$. Consider $z, z^{\prime} \in \mathcal{B}$. A feasible path from $z$ to $z^{\prime}$ consisting of multiple feasible transitions of $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is denoted by $z \Rightarrow z^{\prime}$. The reachable set from $z$ is denoted as $\diamond z:=\left\{z^{\prime} \in \mathcal{B} \mid z \Rightarrow z^{\prime}\right\}$.

Lemma 4.1. $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is a regular perturbation of $\left\{\mathcal{P}_{t}^{0}\right\}$.
Proof. Consider a feasible transition $z^{1} \rightarrow z^{2}$ with $z^{1}:=\left(s^{0}, s^{1}\right)$ and $z^{2}:=\left(s^{1}, s^{2}\right)$. Then we can define a partition of $V$ as $\Lambda_{1}:=\left\{i \in V \mid s_{i}^{2}=s_{i}^{\tau_{i}(0,1)}\right\}$ and $\Lambda_{2}:=\{i \in$ $\left.V \mid s_{i}^{2} \in \mathcal{F}_{i}\left(a_{i}^{1}\right) \backslash\left\{s_{i}^{\tau_{i}(0,1)}\right\}\right\}$. The corresponding probability is given by

$$
\begin{equation*}
P_{z^{1} z^{2}}^{\epsilon}=\prod_{i \in \Lambda_{1}}(1-\epsilon) \times \prod_{j \in \Lambda_{2}} \frac{\epsilon}{\left|\mathcal{F}_{i}\left(a_{i}^{1}\right)\right|-1} . \tag{4.1}
\end{equation*}
$$

Hence, the resistance of the transition $z^{1} \rightarrow z^{2}$ is $\left|\Lambda_{2}\right| \in\{0,1, \ldots, N\}$ since

$$
0<\lim _{\epsilon \rightarrow 0^{+}} \frac{P_{z^{1} z^{2}}^{\epsilon}}{\epsilon^{\left|\Lambda_{2}\right|}}=\prod_{j \in \Lambda_{2}} \frac{1}{\left|\mathcal{F}_{i}\left(a_{i}^{1}\right)\right|-1}<+\infty
$$

We have that (A3) in section 7.2 holds. It is not difficult to see that (A2) holds, and we are now in a position to verify (A1). Since $\mathcal{G}_{\text {loc }}$ is undirected and connected, and multiple sensors can stay in the same position, then $\diamond a^{0}=\mathcal{Q}^{N}$ for any $a^{0} \in \mathcal{Q}$.

Since sensor $i$ can choose any camera control vector from $\mathcal{C}$ at each time, then $\diamond s^{0}=\mathcal{A}$ for any $s^{0} \in \mathcal{A}$. It implies that $\diamond z^{0}=\mathcal{B}$ for any $z^{0} \in \mathcal{B}$, and thus the Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is irreducible on the space $\mathcal{B}$.

It is easy to see that any state in $\operatorname{diag} \mathcal{A}$ has period 1 . Pick any $\left(s^{0}, s^{1}\right) \in \mathcal{B} \backslash \operatorname{diag} \mathcal{A}$. Since $\mathcal{G}_{\text {loc }}$ is undirected, then $s_{i}^{0} \in \mathcal{F}_{i}\left(a_{i}^{1}\right)$ if and only if $s_{i}^{1} \in \mathcal{F}_{i}\left(a_{i}^{0}\right)$. Hence, the following two paths are both feasible:

$$
\begin{aligned}
\left(s^{0}, s^{1}\right) & \rightarrow\left(s^{1}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right) \\
\left(s^{0}, s^{1}\right) & \rightarrow\left(s^{1}, s^{1}\right) \rightarrow\left(s^{1}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right)
\end{aligned}
$$

Hence, the period of the state $\left(s^{0}, s^{1}\right)$ is 1 . This proves aperiodicity of $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$. Since $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is irreducible and aperiodic, then (A1) holds.

Lemma 4.2. For any $\left(s^{0}, s^{0}\right) \in \operatorname{diag} \mathcal{A} \backslash \operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$, there is a finite sequence of transitions from $\left(s^{0}, s^{0}\right)$ to some $\left(s^{*}, s^{*}\right) \in \operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$ that satisfies

$$
\begin{gathered}
\mathcal{L}:=\left(s^{0}, s^{0}\right) \xrightarrow{O(\epsilon)}\left(s^{0}, s^{1}\right) \xrightarrow{O(1)}\left(s^{1}, s^{1}\right) \xrightarrow{O(\epsilon)}\left(s^{1}, s^{2}\right) \\
\xrightarrow{O(1)}\left(s^{2}, s^{2}\right) \xrightarrow{O(\epsilon)} \cdots \xrightarrow{O(\epsilon)}\left(s^{k-1}, s^{k}\right) \xrightarrow{O(1)}\left(s^{k}, s^{k}\right),
\end{gathered}
$$

where $\left(s^{k}, s^{k}\right)=\left(s^{*}, s^{*}\right)$ for some $k \geq 1$.
Proof. If $s^{0} \notin \mathcal{E}\left(\Gamma_{\text {cov }}\right)$, there exists a sensor $i$ with an action $s_{i}^{1} \in \mathcal{F}_{i}\left(a_{i}^{0}\right)$ such that $u_{i}\left(s^{1}\right)>u_{i}\left(s^{0}\right)$, where $s_{-i}^{0}=s_{-i}^{1}$. The transition $\left(s^{0}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right)$ happens when only sensor $i$ experiments, and its corresponding probability is $(1-\epsilon)^{N-1} \times \frac{\epsilon}{\left|\mathcal{F}_{i}\left(a_{i}^{0}\right)\right|-1}$. Since the function $\phi$ is the potential function of the game $\Gamma_{\text {cov }}$, then we have that $\phi\left(s^{1}\right)-\phi\left(s^{0}\right)=u_{i}\left(s^{1}\right)-u_{i}\left(s^{0}\right)$ and thus $\phi\left(s^{1}\right)>\phi\left(s^{0}\right)$.

Since $u_{i}\left(s^{1}\right)>u_{i}\left(s^{0}\right)$ and $s_{-i}^{0}=s_{-i}^{1}$, the transition $\left(s^{0}, s^{1}\right) \rightarrow\left(s^{1}, s^{1}\right)$ occurs when all sensors do not experiment, and the associated probability is $(1-\epsilon)^{N}$.

We repeat the above process and construct the path $\mathcal{L}$ with length $k \geq 1$. Since $\phi\left(s^{i}\right)>\phi\left(s^{i-1}\right)$ for $i=\{1, \ldots, k\}$, then $s^{i} \neq s^{j}$ for $i \neq j$ and thus the path $\mathcal{L}$ has no loop. Since $\mathcal{A}$ is finite, then $k$ is finite and thus $s^{k}=s^{*} \in \mathcal{E}\left(\Gamma_{\text {cov }}\right)$.

A direct result of Lemma 4.1 is that for each $\epsilon$, there exists a unique stationary distribution of $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$, say $\mu(\epsilon)$. We now proceed to utilize Theorem 7.6 to characterize $\lim _{\epsilon \rightarrow 0^{+}} \mu(\epsilon)$.

Proposition 4.3. Consider the regular perturbation $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ of $\left\{\mathcal{P}_{t}^{0}\right\}$. Then $\lim _{\epsilon \rightarrow 0^{+}} \mu(\epsilon)$ exists and the limiting distribution $\mu(0)$ is a stationary distribution of $\left\{\mathcal{P}_{t}^{0}\right\}$. Furthermore, the stochastically stable states (i.e., the support of $\mu(0)$ ) are contained in the set $\operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$.

Proof. Notice that the stochastically stable states are contained in the recurrent communication classes of the unperturbed Markov chain that corresponds to the DHSCL algorithm with $\epsilon=0$. Thus the stochastically stable states are included in the set $\operatorname{diag} \mathcal{A} \subset \mathcal{B}$. Denote by $T_{\min }$ the minimum resistance tree and by $h_{v}$ the root of $T_{\min }$. Each edge of $T_{\min }$ has resistance $0,1,2, \ldots$ corresponding to the transition probability $O(1), O(\epsilon), O\left(\epsilon^{2}\right), \ldots$ The state $z^{\prime}$ is the successor of the state $z$ if and only if $\left(z, z^{\prime}\right) \in T_{\text {min }}$. Like Theorem 3.2 in [22], our analysis will be slightly different from the presentation in section 7.2 . We will construct $T_{\text {min }}$ over states in the set $\mathcal{B}$ (rather than $\operatorname{diag} \mathcal{A}$ ) with the restriction that all the edges leaving the states in $\mathcal{B} \backslash \operatorname{diag} \mathcal{A}$ have resistance 0 . The stochastically stable states are not changed under this difference.

Claim 1. For any $\left(s^{0}, s^{1}\right) \in \mathcal{B} \backslash \operatorname{diag} \mathcal{A}$, there is a finite path

$$
\mathcal{L}^{\prime}:=\left(s^{0}, s^{1}\right) \xrightarrow{O(1)}\left(s^{1}, s^{2}\right) \xrightarrow{O(1)}\left(s^{2}, s^{2}\right),
$$

where $s_{i}^{2}=s_{i}^{\tau_{i}(0,1)}$ for all $i \in V$.

Proof. These two transitions occur when all agents do not experiment. The corresponding probability of each transition is $(1-\epsilon)^{N}$.

Claim 2. The root $h_{v}$ belongs to the set $\operatorname{diag} \mathcal{A}$.
Proof. Suppose that $h_{v}=\left(s^{0}, s^{1}\right) \in \mathcal{B} \backslash \operatorname{diag} \mathcal{A}$. By Claim 1, there is a finite path $\mathcal{L}^{\prime}:=\left(s^{0}, s^{1}\right) \xrightarrow{O(1)}\left(s^{1}, s^{2}\right) \xrightarrow{O(1)}\left(s^{2}, s^{2}\right)$. We now construct a new tree $T^{\prime}$ by adding the edges of the path $\mathcal{L}^{\prime}$ into the tree $T_{\min }$ and removing the redundant edges. The total resistance of adding edges is 0 . Observe that the resistance of the removed edge exiting from $\left(s^{2}, s^{2}\right)$ in the tree $T_{\min }$ is at least 1 . Hence, the resistance of $T^{\prime}$ is strictly lower than that of $T_{\min }$, and we get a contradiction.

Claim 3. Pick any $s^{*} \in \mathcal{E}\left(\Gamma_{\text {cov }}\right)$ and consider $z:=\left(s^{*}, s^{*}\right), z^{\prime}:=\left(s^{*}, \tilde{s}\right)$, where $\tilde{s} \neq s^{*}$. If $\left(z, z^{\prime}\right) \in T_{\min }$, then the resistance of the edge $\left(z, z^{\prime}\right)$ is some $k \geq 2$.

Proof. Suppose the deviator in the transition $z \rightarrow z^{\prime}$ is unique, say $i$. Then the corresponding transition probability is $O(\epsilon)$. Since $s^{*} \in \mathcal{E}\left(\Gamma_{\text {cov }}\right)$ and $\tilde{s}_{i} \in \mathcal{F}_{i}\left(a_{i}^{*}\right)$, we have that $u_{i}\left(s_{i}^{*}, s_{-i}^{*}\right) \geq u_{i}\left(\tilde{s}_{i}, \tilde{s}_{-i}\right)$, where $s_{-i}^{*}=\tilde{s}_{-i}$.

Since $z^{\prime} \in \mathcal{B} \backslash \operatorname{diag} \mathcal{A}$, it follows from Claim 2 that the state $z^{\prime}$ cannot be the root of $T_{\min }$ and thus has a successor $z^{\prime \prime}$. Note that all the edges leaving the states in $\mathcal{B} \backslash \operatorname{diag} \mathcal{A}$ have resistance 0 . Then no experiments in the transition $z^{\prime} \rightarrow z^{\prime \prime}$ and $z^{\prime \prime}=(\tilde{s}, \hat{s})$ for some $\hat{s}$. Since $u_{i}\left(s_{i}^{*}, s_{-i}^{*}\right) \geq u_{i}\left(\tilde{s}_{i}, \tilde{s}_{-i}\right)$ with $s_{-i}^{*}=\tilde{s}_{-i}$, we have $\hat{s}=s^{*}$ and thus $z^{\prime \prime}=\left(\tilde{s}, s^{*}\right)$. Similarly, the state $z^{\prime \prime}$ must have a successor $z^{\prime \prime \prime}$ and $z^{\prime \prime \prime}=z$. We then obtain a loop in $T_{\min }$ which contradicts that $T_{\min }$ is a tree.

It implies that at least two sensors experiment in the transition $z \rightarrow z^{\prime}$. Thus the resistance of the edge $\left(z, z^{\prime}\right)$ is at least 2 .

Claim 4. The root $h_{v}$ belongs to the set $\operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$.
Proof. Suppose that $h_{v}=\left(s^{0}, s^{0}\right) \notin \operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$. By Lemma 4.2, there is a finite path $\mathcal{L}$ connecting $\left(s^{0}, s^{0}\right)$ and some $\left(s^{*}, s^{*}\right) \in \operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$. We now construct a new tree $T^{\prime}$ by adding the edges of the path $\mathcal{L}$ into the tree $T_{\text {min }}$ and removing the edges that leave the states in $\mathcal{L}$ in the tree $T_{\min }$. The total resistance of adding edges is $k$. Observe that the resistance of the removed edge exiting from $\left(s^{i}, s^{i}\right)$ in the tree $T_{\min }$ is at least 1 for $i \in\{1, \ldots, k-1\}$. By Claim 3, the resistance of the removed edge leaving from $\left(s^{*}, s^{*}\right)$ in the tree $T_{\min }$ is at least 2. The total resistance of removing edges is at least $k+1$. Hence, the resistance of $T^{\prime}$ is strictly lower than that of $T_{\min }$, and we get a contradiction.

It follows from Claim 4 that the states in $\operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$ have minimum stochastic potential. Since Lemma 4.1 shows that Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is a regularly perturbed Markov process, Proposition 4.3 is a direct result of Theorem 7.6.

We are now ready to show the proof of Theorem 3.1.
Proof of Theorem 3.1.
Claim 5. Condition (B2) in Theorem 7.5 holds.
Proof. For each $t \geq 0$ and each $z \in X$, we define the numbers

$$
\begin{aligned}
& \sigma_{z}(\epsilon(t)):=\sum_{T \in G(z)} \prod_{(x, y) \in T} P_{x y}^{\epsilon(t)}, \quad \sigma_{z}^{t}=\sigma_{z}(\epsilon(t)), \\
& \mu_{z}(\epsilon(t)):=\frac{\sigma_{z}(\epsilon(t))}{\sum_{x \in X} \sigma_{x}(\epsilon(t))}, \quad \mu_{z}^{t}=\mu_{z}(\epsilon(t))
\end{aligned}
$$

Since $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is a regular perturbation of $\left\{\mathcal{P}_{t}^{0}\right\}$, then it is irreducible and thus $\sigma_{z}^{t}>0$. As Lemma 3.1 of Chapter 6 in [10], one can show that $\left(\mu^{t}\right)^{T} P^{\epsilon(t)}=\left(\mu^{t}\right)^{T}$. Therefore, condition (B2) in Theorem 7.5 holds.

Claim 6. Condition (B3) in Theorem 7.5 holds.
Proof. We now proceed to verify condition (B3) in Theorem 7.5. To do that, let us first fix $t$, denote $\epsilon=\epsilon(t)$, and study the monotonicity of $\mu_{z}(\epsilon)$ with respect to $\epsilon$. We write $\sigma_{z}(\epsilon)$ in the form

$$
\begin{equation*}
\sigma_{z}(\epsilon)=\sum_{T \in G(z)} \prod_{(x, y) \in T} P_{x y}^{\epsilon}=\sum_{T \in G(z)} \prod_{(x, y) \in T} \frac{\alpha_{x y}(\epsilon)}{\beta_{x y}(\epsilon)}=\frac{\alpha_{z}(\epsilon)}{\beta_{z}(\epsilon)} \tag{4.2}
\end{equation*}
$$

for some polynomials $\alpha_{z}(\epsilon)$ and $\beta_{z}(\epsilon)$ in $\epsilon$. With (4.2) in hand, we have that $\sum_{x \in X}$ $\sigma_{x}(\epsilon)$ and thus $\mu_{z}(\epsilon)$ are ratios of two polynomials in $\epsilon$, i.e., $\mu_{z}(\epsilon)=\frac{\varphi_{z}(\epsilon)}{\beta(\epsilon)}$, where $\varphi_{z}(\epsilon)$ and $\beta(\epsilon)$ are polynomials in $\epsilon$. The derivative of $\mu_{z}(\epsilon)$ is given by

$$
\frac{\partial \mu_{z}(\epsilon)}{\partial \epsilon}=\frac{1}{\beta(\epsilon)^{2}}\left(\frac{\partial \varphi_{z}(\epsilon)}{\partial \epsilon} \beta(\epsilon)-\varphi_{z}(\epsilon) \frac{\partial \beta(\epsilon)}{\partial \epsilon}\right) .
$$

Note that the numerator $\frac{\partial \varphi_{z}(\epsilon)}{\partial \epsilon} \beta(\epsilon)-\varphi_{z}(\epsilon) \frac{\partial \beta(\epsilon)}{\partial \epsilon}$ is a polynomial in $\epsilon$. Denote by $\iota_{z} \neq 0$ the coefficient of the leading term of $\frac{\partial \varphi_{z}(\epsilon)}{\partial \epsilon}-\varphi_{z}(\epsilon) \frac{\partial \beta(\epsilon)}{\epsilon}$. The leading term dominates $\frac{\partial \varphi_{z}(\epsilon)}{\partial \epsilon}-\varphi_{z}(\epsilon) \frac{\partial \beta(\epsilon)}{\epsilon}$ when $\epsilon$ is sufficiently small. Thus there exists $\epsilon_{z}>0$ such that the sign of $\frac{\partial \mu_{z}(\epsilon)}{\partial \epsilon}$ is the sign of $\iota_{z}$ for all $0<\epsilon \leq \epsilon_{z}$. Let $\epsilon^{*}=\max _{z \in X} \epsilon_{z}$.

Since $\epsilon(t)$ strictly decreases to zero, there is a unique finite time instant $t^{*}$ such that $\epsilon\left(t^{*}\right)=\epsilon^{*}$ (if $\epsilon(0)<\epsilon^{*}$, then $t^{*}=0$ ). Since $\epsilon(t)$ is strictly decreasing, we can define a partition of $X$ as follows:

$$
\begin{aligned}
& \Xi_{1}:=\left\{z \in X \mid \mu_{z}(\epsilon(t))>\mu_{z}(\epsilon(t+1)) \quad \forall t \in\left[t^{*},+\infty\right)\right\}, \\
& \Xi_{2}:=\left\{z \in X \mid \mu_{z}(\epsilon(t))<\mu_{z}(\epsilon(t+1)) \quad \forall t \in\left[t^{*},+\infty\right)\right\} .
\end{aligned}
$$

We are now ready to verify (B3) of Theorem 7.5. Since $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is a regular perturbed Markov chain of $\left\{\mathcal{P}_{t}^{0}\right\}$, it follows from Theorem 7.6 that $\lim _{t \rightarrow+\infty} \mu_{z}(\epsilon(t))=\mu_{z}(0)$, and thus it holds that

$$
\begin{aligned}
& \sum_{t=0}^{+\infty} \sum_{z \in X}\left\|\mu_{z}^{t}-\mu_{z}^{t+1}\right\|=\sum_{t=0}^{+\infty} \sum_{z \in X}\left|\mu_{z}(\epsilon(t))-\mu_{z}(\epsilon(t+1))\right| \\
& =\sum_{t=0}^{t^{*}} \sum_{z \in X}\left|\mu_{z}(\epsilon(t))-\mu_{z}(\epsilon(t+1))\right|+\sum_{t=t^{*}+1}^{+\infty}\left(\sum_{z \in \Xi_{1}} \mu_{z}(\epsilon(t))-\sum_{z \in \Xi_{1}} \mu_{z}(\epsilon(t+1))\right) \\
& +\sum_{t=t^{*}+1}^{+\infty}\left(1-\sum_{z \in \Xi_{1}} \mu_{z}(\epsilon(t+1))-\left(1-\sum_{z \in \Xi_{1}} \mu_{z}(\epsilon(t))\right)\right) \\
& =\sum_{t=0}^{t^{*}} \sum_{z \in X}\left|\mu_{z}(\epsilon(t))-\mu_{z}(\epsilon(t+1))\right|+2 \sum_{z \in \Xi_{1}} \mu_{z}\left(\epsilon\left(t^{*}+1\right)\right)-2 \sum_{z \in \Xi_{1}} \mu_{z}(0)<+\infty .
\end{aligned}
$$

Claim 7. Condition (B1) in Theorem 7.5 holds.
Proof. Denote by $P^{\epsilon(t)}$ the transition matrix of $\left\{\mathcal{P}_{t}\right\}$. As in (4.1), the probability of the feasible transition $z^{1} \rightarrow z^{2}$ is given by

$$
P_{z^{1} z^{2}}^{\epsilon(t)}=\prod_{i \in \Lambda_{1}}(1-\epsilon(t)) \times \prod_{j \in \Lambda_{2}} \frac{\epsilon(t)}{\left|\mathcal{F}_{i}\left(a_{i}^{1}\right)\right|-1} .
$$

Observe that $\left|\mathcal{F}_{i}\left(a_{i}^{1}\right)\right| \leq 5|\mathcal{C}|$. Since $\epsilon(t)$ is strictly decreasing, there is $t_{0} \geq 1$ such that $t_{0}$ is the first time when $1-\epsilon(t) \geq \frac{\epsilon(t)}{5|\mathcal{C}|-1}$. Then for all $t \geq t_{0}$ it holds that

$$
P_{z^{1} z^{2}}^{\epsilon(t)} \geq\left(\frac{\epsilon(t)}{5|\mathcal{C}|-1}\right)^{N}
$$

Denote $P(m, n):=\prod_{t=m}^{n-1} P^{\epsilon(t)}, 0 \leq m<n$. Pick any $z \in \mathcal{B}$ and let $u_{z} \in \mathcal{B}$ be such that $P_{u_{z} z}(t, t+D+1)=\min _{x \in \mathcal{B}} P_{x z}(t, t+D+1)$. Consequently, it follows that for all $t \geq t_{0}$

$$
\begin{aligned}
& \min _{x \in \mathcal{B}} P_{x z}(t, t+D+1)=\sum_{i_{1} \in \mathcal{B}} \cdots \sum_{i_{D} \in \in \mathcal{B}} P_{u_{z} i_{1}}^{\epsilon(t)} \cdots P_{i_{D-1} i_{D}}^{\epsilon(t+D-1)} P_{i_{D} z}^{\epsilon(t+D)} \\
& \geq P_{u_{z} i_{1}}^{\epsilon(t)} \cdots P_{i_{D-1} i_{D}}^{\epsilon(t+D-1)} P_{i_{D} z}^{\epsilon(t+D)} \geq \prod_{i=0}^{D}\left(\frac{\epsilon(t+i)}{5|\mathcal{C}|-1}\right)^{N} \geq\left(\frac{\epsilon(t)}{5|\mathcal{C}|-1}\right)^{(D+1) N}
\end{aligned}
$$

where in the last inequality we use that $\epsilon(t)$ is strictly decreasing. Then we have

$$
\begin{aligned}
& 1-\lambda(P(t, t+D+1))=\min _{x, y \in \mathcal{B}} \sum_{z \in \mathcal{B}} \min \left\{P_{x z}(t, t+D+1), P_{y z}(t, t+D+1)\right\} \\
& \geq \sum_{z \in \mathcal{B}} P_{u_{z} z}(t, t+D+1) \geq|\mathcal{B}|\left(\frac{\epsilon(t)}{5|\mathcal{C}|-1}\right)^{(D+1) N}
\end{aligned}
$$

Choose $k_{i}:=(D+1) i$ and let $i_{0}$ be the smallest integer such that $(D+1) i_{0} \geq t_{0}$. Then we have that

$$
\begin{align*}
& \sum_{i=0}^{+\infty}\left(1-\lambda\left(P\left(k_{i}, k_{i+1}\right)\right)\right) \geq|\mathcal{B}| \sum_{i=i_{0}}^{+\infty}\left(\frac{\epsilon((D+1) i)}{5|\mathcal{C}|-1}\right)^{(D+1) N} \\
& =\frac{|\mathcal{B}|}{(5|\mathcal{C}|-1)^{(D+1) N}} \sum_{i=i_{0}}^{+\infty} \frac{1}{(D+1) i}=+\infty \tag{4.3}
\end{align*}
$$

Hence, the weak ergodicity property follows from Theorem 7.4.
All the conditions in Theorem 7.5 hold. Thus it follows from Theorem 7.5 that the limiting distribution is $\mu^{*}=\lim _{t \rightarrow+\infty} \mu^{t}$. Note that $\lim _{t \rightarrow+\infty} \mu^{t}=\lim _{t \rightarrow+\infty} \mu(\epsilon(t))=$ $\mu(0)$ and Proposition 4.3 shows that the support of $\mu(0)$ is contained in the set $\operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$. Hence, the support of $\mu^{*}$ is contained in the set $\operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)$, implying that $\lim _{t \rightarrow+\infty} \mathbb{P}\left(z(t) \in \operatorname{diag} \mathcal{E}\left(\Gamma_{\text {cov }}\right)\right)=1$. This completes the proof.
4.2. Convergence analysis of the DIACL Algorithm. First of all, we employ Theorem 7.6 to study the convergence properties of the associated DHACL algorithm. This is essential to analyzing the DIACL algorithm.

To simplify notation, we will use $s_{i}(t-1):=s_{i}\left(\gamma_{i}^{(2)}(t)+1\right)$ in the remainder of this section. Observe that $z(t):=(s(t-1), s(t))$ in the DHACL algorithm constitutes a Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ on the space $\mathcal{B}^{\prime}$.

Lemma 4.4. The Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is a regular perturbation of $\left\{\mathcal{P}_{t}^{0}\right\}$.
Proof. Pick any two states $z^{1}:=\left(s^{0}, s^{1}\right)$ and $z^{2}:=\left(s^{1}, s^{2}\right)$ with $z^{1} \neq z^{2}$. We have that $P_{z^{1} z^{2}}^{\epsilon}>0$ if and only if there is some $i \in V$ such that $s_{-i}^{1}=s_{-i}^{2}$ and one of the following occurs: $s_{i}^{2} \in \mathcal{F}_{i}\left(a_{i}^{1}\right) \backslash\left\{s_{i}^{0}, s_{i}^{1}\right\}, s_{i}^{2}=s_{i}^{1}$ or $s_{i}^{2}=s_{i}^{0}$. In particular, the
following holds:

$$
P_{z^{1} z^{2}}^{\epsilon}= \begin{cases}\eta_{1}, & s_{i}^{2} \in \mathcal{F}_{i}\left(a_{i}^{1}\right) \backslash\left\{s_{i}^{0}, s_{i}^{1}\right\} \\ \eta_{2}, & s_{i}^{2}=s_{i}^{1} \\ \eta_{3}, & s_{i}^{2}=s_{i}^{0}\end{cases}
$$

where

$$
\eta_{1}:=\frac{\epsilon^{m_{i}}}{N\left|\mathcal{F}_{i}\left(a_{i}^{1}\right) \backslash\left\{s_{i}^{0}, s_{i}^{1}\right\}\right|}, \quad \eta_{2}:=\frac{1-\epsilon^{m_{i}}}{N\left(1+\epsilon^{\rho_{i}\left(s^{0}, s^{1}\right)}\right)}, \quad \eta_{3}:=\frac{\left.\left(1-\epsilon^{m_{i}}\right) \times \epsilon^{\rho_{i}\left(s^{0}, s^{1}\right.}\right)}{N\left(1+\epsilon^{\rho_{i}\left(s^{0}, s^{1}\right)}\right)}
$$

Observe that $0<\lim _{\epsilon \rightarrow 0^{+}} \frac{\eta_{1}}{\epsilon^{m_{i}}}<+\infty$. Multiplying the numerator and denominator of $\eta_{2}$ by $\epsilon^{\Psi_{i}\left(s^{1}, s^{0}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right)}$, we obtain

$$
\eta_{2}=\frac{1-\epsilon^{m_{i}}}{N} \times \frac{\epsilon^{\Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right)}}{\eta_{2}^{\prime}}
$$

where $\eta_{2}^{\prime}:=\epsilon^{\Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right)}+\epsilon^{\Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right)\right)}$. Use

$$
\lim _{\epsilon \rightarrow 0^{+}} \epsilon^{x}= \begin{cases}1, & x=0 \\ 0, & x>0\end{cases}
$$

and we have

$$
\lim _{\epsilon \rightarrow 0^{+}} \frac{\eta_{2}}{\epsilon^{\Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right)}}= \begin{cases}\frac{1}{N}, & u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right) \neq u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right) \\ \frac{1}{2 N} & \text { otherwise }\end{cases}
$$

Similarly, it holds that

$$
\lim _{\epsilon \rightarrow 0^{+}} \frac{\eta_{3}}{\epsilon^{\Psi i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right)\right)} \in\left\{\frac{1}{2 N}, \frac{1}{N}\right\} .
$$

Hence, the resistance of the feasible transition $z^{1} \rightarrow z^{2}$, with $z^{1} \neq z^{2}$ and sensor $i$ as the unilateral deviator, can be described as follows:

$$
\chi\left(z^{1} \rightarrow z^{2}\right)= \begin{cases}m_{i}, \quad s_{i}^{2} \in \mathcal{F}_{i}\left(a^{1}\right) \backslash\left\{s_{i}^{0}, s_{i}^{1}\right\}, \\ \Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right), & s_{i}^{2}=s_{i}^{1} \\ \Psi_{i}\left(s^{0}, s^{1}\right)-\left(u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right)\right), & s_{i}^{2}=s_{i}^{0}\end{cases}
$$

Then (A3) in section 7.2 holds. It is straightforward to verify that (A2) in section 7.2 holds. We are now in a position to verify (A1). Since $\mathcal{G}_{\text {loc }}$ is undirected and connected, and multiple sensors can stay in the same position, then $\diamond a^{0}=\mathcal{Q}^{N}$ for any $a^{0} \in \mathcal{Q}$. Since sensor $i$ can choose any camera control vector from $\mathcal{C}$ at each time, then $\diamond s^{0}=\mathcal{A}$ for any $s^{0} \in \mathcal{A}$. This implies that $\diamond z^{0}=\mathcal{B}^{\prime}$ for any $z^{0} \in \mathcal{B}^{\prime}$, and thus the Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is irreducible on the space $\mathcal{B}^{\prime}$.

It is easy to see that any state in $\operatorname{diag} \mathcal{A}$ has period 1 . Pick any $\left(s^{0}, s^{1}\right) \in$ $\mathcal{B}^{\prime} \backslash \operatorname{diag} \mathcal{A}$. Since $\mathcal{G}_{\text {loc }}$ is undirected, then $s_{i}^{0} \in \mathcal{F}_{i}\left(a_{i}^{1}\right)$ if and only if $s_{i}^{1} \in \mathcal{F}_{i}\left(a_{i}^{0}\right)$. Hence, the following two paths are both feasible:

$$
\begin{aligned}
& \left(s^{0}, s^{1}\right) \rightarrow\left(s^{1}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right) \\
& \left(s^{0}, s^{1}\right) \rightarrow\left(s^{1}, s^{1}\right) \rightarrow\left(s^{1}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right)
\end{aligned}
$$

Hence, the period of the state $\left(s^{0}, s^{1}\right)$ is 1 . This proves aperiodicity of $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$. Since $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is irreducible and aperiodic, then (A1) holds.

A direct result of Lemma 4.4 is that for each $\epsilon>0$, there exists a unique stationary distribution of $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$, say $\mu(\epsilon)$. From the proof of Lemma 4.4, we can see that the resistance of an experiment is $m_{i}$ if sensor $i$ is the unilateral deviator. We now proceed to utilize Theorem 7.6 to characterize $\lim _{\epsilon \rightarrow 0^{+}} \mu(\epsilon)$.

Proposition 4.5. Consider the regular perturbed Markov process $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$. Then $\lim _{\epsilon \rightarrow 0^{+}} \mu(\epsilon)$ exists and the limiting distribution $\mu(0)$ is a stationary distribution of $\left\{\mathcal{P}_{t}^{0}\right\}$. Furthermore, the stochastically stable states (i.e., the support of $\mu(0)$ ) are contained in the set diag $S^{*}$.

Proof. The unperturbed Markov chain corresponds to the DHACL algorithm with $\epsilon=0$. Hence, the recurrent communication classes of the unperturbed Markov chain are contained in the set $\operatorname{diag} \mathcal{A}$. We will construct resistance trees over vertices in the set $\operatorname{diag} \mathcal{A}$. Denote by $T_{\min }$ the minimum resistance tree. The remainder of the proof is divided into the following four claims.

CLAim 8. $\chi\left(\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{1}, s^{1}\right)\right)=m_{i}+\Psi_{i}\left(s^{1}, s^{0}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right)$, where $s^{0} \neq s^{1}$ and the transition $s^{0} \rightarrow s^{1}$ is feasible with sensor $i$ as the unilateral deviator.

Proof. One feasible path for $\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{1}, s^{1}\right)$ is $\mathcal{L}:=\left(s^{0}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right) \rightarrow$ $\left(s^{1}, s^{1}\right)$, where sensor $i$ experiments in the first transition and does not experiment in the second one. The total resistance of the path $\mathcal{L}$ is $m_{i}+\Psi_{i}\left(s^{1}, s^{0}\right)-\left(u_{i}\left(s^{1}\right)-\right.$ $\left.\Delta_{i}\left(s^{1}, s^{0}\right)\right)$, which is at most $m_{i}+m^{*}$.

Denote by $\mathcal{L}^{\prime}$ the path with minimum resistance among all the feasible paths for $\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{1}, s^{1}\right)$. Assume that the first transition in $\mathcal{L}^{\prime}$ is $\left(s^{0}, s^{0}\right) \rightarrow\left(s^{0}, s^{2}\right)$, where node $j$ experiments and $s^{2} \neq s^{1}$. Observe that the resistance of $\left(s^{0}, s^{0}\right) \rightarrow\left(s^{0}, s^{2}\right)$ is $m_{j}$. No matter whether $j$ is equal to $i$ or not, the path $\mathcal{L}^{\prime}$ must include at least one more experiment to introduce $s_{i}^{1}$. Hence the total resistance of the path $\mathcal{L}^{\prime}$ is at least $m_{i}+m_{j}$. Since $m_{i}+m_{j}>m_{i}+2 m^{*}$, the path $\mathcal{L}^{\prime}$ has a strictly larger resistance than the path $\mathcal{L}$. To avoid a contradiction, the path $\mathcal{L}^{\prime}$ must start from the transition $\left(s^{0}, s^{0}\right) \rightarrow\left(s^{0}, s^{1}\right)$. Similarly, the sequent transition (which is also the last one) in the path $\mathcal{L}^{\prime}$ must be $\left(s^{0}, s^{1}\right) \rightarrow\left(s^{1}, s^{1}\right)$ and thus $\mathcal{L}^{\prime}=\mathcal{L}$. Hence, the resistance of the transition $\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{1}, s^{1}\right)$ is the total resistance of the path $\mathcal{L}$, i.e., $m_{i}+\Psi_{i}\left(s^{1}, s^{0}\right)-\left(u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)\right)$.

Claim 9. All the edges $\left((s, s),\left(s^{\prime}, s^{\prime}\right)\right)$ in $T_{\text {min }}$ must consist of only one deviator, i.e., $s_{i} \neq s_{i}^{\prime}$ and $s_{-i}=s_{-i}^{\prime}$ for some $i \in V$.

Proof. Assume that $(s, s) \Rightarrow\left(s^{\prime}, s^{\prime}\right)$ has at least two deviators. Suppose the path $\hat{\mathcal{L}}$ has the minimum resistance among all the paths from $(s, s)$ to $\left(s^{\prime}, s^{\prime}\right)$. Then $\ell \geq 2$ experiments are carried out along $\hat{\mathcal{L}}$. Denote by $i_{k}$ the unilateral deviator in the $k$ th experiment $s^{k-1} \rightarrow s^{k}$, where $1 \leq k \leq \ell, s^{0}=s$, and $s^{\ell}=s^{\prime}$. Then the resistance of $\hat{\mathcal{L}}$ is at least $\sum_{k=1}^{\ell} m_{i_{k}}$, i.e., $\chi\left(\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{\prime}, s^{\prime}\right)\right) \geq \sum_{k=1}^{\ell} m_{i_{k}}$.

Let us consider the following path on $T_{\text {min }}$ :

$$
\overline{\mathcal{L}}:=\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{1}, s^{1}\right) \Rightarrow \cdots \Rightarrow\left(s^{\ell}, s^{\ell}\right)
$$

From Claim 1, we know that the total resistance of the path $\overline{\mathcal{L}}$ is at most $\sum_{k=1}^{\ell} m_{i_{k}}+$ $\ell m^{*}$.

A new tree $T^{\prime}$ can be obtained by adding the edges of $\overline{\mathcal{L}}$ into $T_{\text {min }}$ and removing the redundant edges. The removed resistance is strictly greater than $\sum_{k=1}^{\ell} m_{i_{k}}+$ $2(\ell-1) m^{*}$, where $\sum_{k=1}^{\ell} m_{i_{k}}$ is the lower bound on the resistance on the edge from $\left(s^{0}, s^{0}\right)$ to $\left(s^{\ell}, s^{\ell}\right)$, and $2(\ell-1) m^{*}$ is the strictly lower bound on the total resistances of leaving $\left(s^{k}, s^{k}\right)$ for $k=1, \ldots, \ell-1$. The added resistance is the total resistance
of $\overline{\mathcal{L}}$, which is at most $\sum_{k=1}^{\ell} m_{i_{k}}+\ell m^{*}$. Since $\ell \geq 2$, we have that $2(\ell-1) m^{*} \geq \ell m^{*}$, and thus $T^{\prime}$ has a strictly lower resistance than $T_{\min }$. This contradicts the fact that $T_{\text {min }}$ is a minimum resistance tree. $\quad \square$

Claim 10. Given any edge $\left((s, s),\left(s^{\prime}, s^{\prime}\right)\right)$ in $T_{\text {min }}$, denote by $i$ the unilateral deviator between $s$ and $s^{\prime}$. Then the transition $s_{i} \rightarrow s_{i}^{\prime}$ is feasible.

Proof. Assume that the transition $s_{i} \rightarrow s_{i}^{\prime}$ is infeasible. Suppose the path $\check{\mathcal{L}}$ has the minimum resistance among all the paths from $(s, s)$ to $\left(s^{\prime}, s^{\prime}\right)$. Then there are $\ell \geq 2$ experiments in $\check{\mathcal{L}}$. The remainder of the proof is similar to that of Claim 9 .

Claim 11. Let $h_{v}$ be the root of $T_{\min }$. Then $h_{v} \in \operatorname{diag} S^{*}$.
Proof. Assume that $h_{v}=\left(s^{0}, s^{0}\right) \notin \operatorname{diag} S^{*}$. Pick any $\left(s^{*}, s^{*}\right) \in \operatorname{diag} S^{*}$. By Claims 9 and 10, we have that there is a path from $\left(s^{*}, s^{*}\right)$ to $\left(s^{0}, s^{0}\right)$ in the tree $T_{\text {min }}$ as follows:

$$
\tilde{\mathcal{L}}:=\left(s^{\ell}, s^{\ell}\right) \Rightarrow\left(s^{\ell-1}, s^{\ell-1}\right) \Rightarrow \cdots \Rightarrow\left(s^{1}, s^{1}\right) \Rightarrow\left(s^{0}, s^{0}\right)
$$

for some $\ell \geq 1$. Here, $s^{*}=s^{\ell}$, there is only one deviator, say $i_{k}$, from $s^{k}$ to $s^{k-1}$, and the transition $s^{k} \rightarrow s^{k-1}$ is feasible for $k=\ell, \ldots, 1$.

Since the transition $s^{k} \rightarrow s^{k+1}$ is also feasible for $k=0, \ldots, \ell-1$, we obtain the reverse path $\tilde{\mathcal{L}}^{\prime}$ of $\tilde{\mathcal{L}}$ as follows:

$$
\tilde{\mathcal{L}}^{\prime}:=\left(s^{0}, s^{0}\right) \Rightarrow\left(s^{1}, s^{1}\right) \Rightarrow \cdots \Rightarrow\left(s^{\ell-1}, s^{\ell-1}\right) \Rightarrow\left(s^{\ell}, s^{\ell}\right)
$$

By Claim 8, the total resistance of the path $\tilde{\mathcal{L}}$ is

$$
\chi(\tilde{\mathcal{L}})=\sum_{k=1}^{\ell} m_{i_{k}}+\sum_{k=1}^{\ell}\left\{\Psi_{i_{k}}\left(s^{k}, s^{k-1}\right)-\left(u_{i_{k}}\left(s^{k-1}\right)-\Delta_{i_{k}}\left(s^{k-1}, s^{k}\right)\right)\right\}
$$

and the total resistance of the path $\tilde{\mathcal{L}}^{\prime}$ is

$$
\chi\left(\tilde{\mathcal{L}}^{\prime}\right)=\sum_{k=1}^{\ell} m_{i_{k}}+\sum_{k=1}^{\ell} \Psi_{i_{k}}\left(s^{k-1}, s^{k}\right)-\left(u_{i_{k}}\left(s^{k}\right)-\Delta_{i_{k}}\left(s^{k}, s^{k-1}\right)\right)
$$

Denote $\Lambda_{1}^{\prime}:=\left(\mathcal{D}\left(a_{i_{k}}^{k}, r_{i_{k}}^{k}\right) \backslash \mathcal{D}\left(a_{i_{k-1}}^{k-1}, r_{i_{k-1}}^{k-1}\right)\right) \cap \mathcal{Q}$ and $\Lambda_{2}^{\prime}:=\left(\mathcal{D}\left(a_{i_{k-1}}^{k-1}, r_{i_{k-1}}^{k-1}\right) \backslash \mathcal{D}\left(a_{i_{k}}^{k}\right.\right.$, $\left.\left.r_{i_{k}}^{k}\right)\right) \cap \mathcal{Q}$. Observe that

$$
\begin{aligned}
& U_{g}\left(s^{k}\right)-U_{g}\left(s^{k-1}\right) \\
& =u_{i_{k}}\left(s^{k}\right)-u_{i_{k}}\left(s^{k-1}\right)-\sum_{q \in \Lambda_{1}^{\prime}} W_{q}\left(\frac{n_{q}\left(s^{k-1}\right)}{n_{q}\left(s^{k-1}\right)}-\frac{n_{q}\left(s^{k-1}\right)}{n_{q}\left(s^{k}\right)}\right) \\
& \quad+\sum_{q \in \Lambda_{2}^{\prime}} W_{q}\left(\frac{n_{q}\left(s^{k}\right)}{n_{q}\left(s^{k}\right)}-\frac{n_{q}\left(s^{k}\right)}{n_{q}\left(s^{k-1}\right)}\right) \\
& =\left(u_{i_{k}}\left(s^{k}\right)-\Delta_{i_{k}}\left(s^{k}, s^{k-1}\right)\right)-\left(u_{i_{k}}\left(s^{k-1}\right)-\Delta_{i_{k}}\left(s^{k-1}, s^{k}\right)\right)
\end{aligned}
$$

We now construct a new tree $T^{\prime}$ with the root $\left(s^{*}, s^{*}\right)$ by adding the edges of $\tilde{\mathcal{L}}^{\prime}$ to the tree $T_{\text {min }}$ and removing the redundant edges $\tilde{\mathcal{L}}$. Since $\Psi_{i_{k}}\left(s^{k-1}, s^{k}\right)=$ $\Psi_{i_{k}}\left(s^{k}, s^{k-1}\right)$, the difference in the total resistances across $\chi\left(T^{\prime}\right)$ and $\chi\left(T_{\min }\right)$ is

$$
\begin{aligned}
& \chi\left(T^{\prime}\right)-\chi\left(T_{\min }\right)=\chi\left(\tilde{\mathcal{L}}^{\prime}\right)-\chi(\tilde{\mathcal{L}}) \\
& =\sum_{k=1}^{\ell}-\left(u_{i_{k}}\left(s^{k-1}\right)-\Delta_{i_{k}}\left(s^{k-1}, s^{k}\right)\right)-\sum_{k=1}^{\ell}-\left(u_{i_{k}}\left(s^{k}\right)-\Delta_{i_{k}}\left(s^{k}, s^{k-1}\right)\right) \\
& =\sum_{k=1}^{\ell}\left(U_{g}\left(s^{k}\right)-U_{g}\left(s^{k-1}\right)\right)=U_{g}\left(s^{0}\right)-U_{g}\left(s^{*}\right)<0 .
\end{aligned}
$$

This contradicts that $T_{\min }$ is a minimum resistance tree. $\quad$
It follows from Claim 4 that the state $h_{v} \in \operatorname{diag} S^{*}$ has minimum stochastic potential. Then Proposition 4.5 is a direct result of Theorem 7.6.

We are now ready to show the proof of Theorem 3.2.
Proof of Theorem 3.2.
Claim 12. Condition (B2) in Theorem 7.5 holds.
Proof. The proof is analogous to Claim 5.
Claim 13. Condition (B3) in Theorem 7.5 holds.
Proof. Denote by $P^{\epsilon(t)}$ the transition matrix of $\left\{\mathcal{P}_{t}\right\}$. Consider the feasible transition $z^{1} \rightarrow z^{2}$ with unilateral deviator $i$. The corresponding probability is given by

$$
P_{z^{1} z^{2}}^{\epsilon(t)}= \begin{cases}\eta_{1}, & s_{i}^{2} \in \mathcal{F}_{i}\left(a_{i}^{1}\right) \backslash\left\{s_{i}^{0}, s_{i}^{1}\right\}, \\ \eta_{2}, & s_{i}^{2}=s_{i}^{1}, \\ \eta_{3}, & s_{i}^{2}=s_{i}^{0},\end{cases}
$$

where

$$
\begin{gathered}
\eta_{1}:=\frac{\epsilon(t)^{m_{i}}}{N\left|\mathcal{F}_{i}\left(a_{i}^{1}\right) \backslash\left\{s_{i}^{0}, s_{i}^{1}\right\}\right|}, \quad \eta_{2}:=\frac{1-\epsilon(t)^{m_{i}}}{N\left(1+\epsilon(t)^{\rho_{i}\left(s^{0}, s^{1}\right)}\right)}, \\
\eta_{3}:=\frac{\left(1-\epsilon(t)^{m_{i}}\right) \times \epsilon(t) \rho_{i}\left(s^{0}, s^{1}\right)}{N\left(1+\epsilon(t)^{\rho_{i}\left(s^{0}, s^{1}\right)}\right)} .
\end{gathered}
$$

The remainder is analogous to Claim $6 . \quad \square$
Claim 14. Condition (B1) in Theorem 7.5 holds.
Proof. Observe that $\left|\mathcal{F}_{i}\left(a_{i}^{1}\right)\right| \leq 5|\mathcal{C}|$. Since $\epsilon(t)$ is strictly decreasing, there is $t_{0} \geq 1$ such that $t_{0}$ is the first time when $1-\epsilon(t)^{m_{i}} \geq \epsilon(t)^{m_{i}}$.

Observe that for all $t \geq 1$, it holds that

$$
\eta_{1} \geq \frac{\epsilon(t)^{m_{i}}}{N(5|\mathcal{C}|-1)} \geq \frac{\epsilon(t)^{m_{i}+m^{*}}}{N(5|\mathcal{C}|-1)}
$$

Denote $b:=u_{i}\left(s^{1}\right)-\Delta_{i}\left(s^{1}, s^{0}\right)$ and $a:=u_{i}\left(s^{0}\right)-\Delta_{i}\left(s^{0}, s^{1}\right)$. Then $\rho_{i}\left(s^{0}, s^{1}\right)=b-a$. Since $b-a \leq m^{*}$, then for $t \geq t_{0}$ it holds that

$$
\begin{aligned}
& \eta_{2}=\frac{1-\epsilon(t)^{m_{i}}}{N\left(1+\epsilon(t)^{b-a}\right)}=\frac{\left(1-\epsilon(t)^{m_{i}}\right) \epsilon(t)^{\max \{a, b\}-b}}{N\left(\epsilon(t)^{\max \{a, b\}-b}+\epsilon(t)^{\max \{a, b\}-a}\right)} \\
& \geq \frac{\epsilon(t)^{m_{i}} \epsilon(t)^{\max \{a, b\}-b}}{2 N} \geq \frac{\epsilon(t)^{m_{i}+m^{*}}}{N(5|\mathcal{C}|-1)} .
\end{aligned}
$$

Similarly, for $t \geq t_{0}$, it holds that

$$
\eta_{3}=\frac{\left(1-\epsilon(t)^{m_{i}}\right) \epsilon(t)^{\max \{a, b\}-a}}{N\left(\epsilon(t)^{\max \{a, b\}-b}+\epsilon(t)^{\max \{a, b\}-a}\right)} \geq \frac{\epsilon(t)^{m_{i}+m^{*}}}{N(5|\mathcal{C}|-1)} .
$$

Since $m_{i} \in\left(2 m^{*}, K m^{*}\right]$ for all $i \in V$ and $K m^{*}>1$, then for any feasible transition $z^{1} \rightarrow z^{2}$ with $z^{1} \neq z^{2}$, it holds that

$$
P_{z^{1} z^{2}}^{\epsilon(t)} \geq \frac{\epsilon(t)^{(K+1) m^{*}}}{N(5|\mathcal{C}|-1)}
$$

for all $t \geq t_{0}$. Furthermore, for all $t \geq t_{0}$ and all $z^{1} \in \operatorname{diag} \mathcal{A}$, we have that

$$
P_{z^{1} z^{1}}^{\epsilon(t)}=1-\frac{1}{N} \sum_{i=1}^{N} \epsilon(t)^{m_{i}}=\frac{1}{N} \sum_{i=1}^{N}\left(1-\epsilon(t)^{m_{i}}\right) \geq \frac{1}{N} \sum_{i=1}^{N} \epsilon(t)^{m_{i}} \geq \frac{\epsilon(t)^{(K+1) m^{*}}}{N(5|\mathcal{C}|-1)} .
$$

Choose $k_{i}:=(D+1) i$ and let $i_{0}$ be the smallest integer such that $(D+1) i_{0} \geq t_{0}$. Similar to (4.3), we can derive the following property:

$$
\sum_{\ell=0}^{+\infty}\left(1-\lambda\left(P\left(k_{\ell}, k_{\ell+1}\right)\right)\right) \geq \frac{|\mathcal{B}|}{(N(5|\mathcal{C}|-1))^{(D+1)(K+1) m^{*}}} \sum_{i=i_{0}}^{+\infty} \frac{1}{(D+1) i}=+\infty .
$$

Hence, the weak ergodicity of $\left\{\mathcal{P}_{t}\right\}$ follows from Theorem 7.4.
All the conditions in Theorem 7.5 hold. Thus it follows from Theorem 7.5 that the limiting distribution is $\mu^{*}=\lim _{t \rightarrow+\infty} \mu^{t}$. Note that $\lim _{t \rightarrow+\infty} \mu^{t}=\lim _{t \rightarrow+\infty} \mu(\epsilon(t))=$ $\mu(0)$ and Proposition 4.5 shows that the support of $\mu(0)$ is contained in the set $\operatorname{diag} S^{*}$. Hence, the support of $\mu^{*}$ is contained in the set diag $S^{*}$, implying that $\lim _{t \rightarrow+\infty} \mathbb{P}\left(z(t) \in \operatorname{diag} S^{*}\right)=1$. This completes the proof.
5. Discussion and simulation. In this section, we present some remarks along with two numerical examples to illustrate the performance of our algorithms.

Theorems 3.1 and 3.2 guarantee the asymptotic convergence in probability of the proposed algorithms. However, our theoretical results do not provide any estimate of the convergence rates, which could be very slow in practice. This is a consequence of the well-known exploration-exploitation trade-off termed in reinforcement learning, e.g., in [32]. Intuitively, each algorithm starts from a relatively large exploration rate, and this allows the algorithm to explore the unknown environment quickly. As time progresses, the exploration rate is decreased, allowing each algorithm to exploit the information collected and converge to some desired configuration. In order to avoid being locked into some undesired configuration, each algorithm requires a very slowly decreasing exploration rate. In the numerical examples below, we have chosen suitable exploration rates empirically.
5.1. A numerical example of the DISCL algorithm. Consider a $10 \times 10$ square in which each grid is $1 \times 1$, and a group of nine mobile sensors are deployed in this area. Note that, given arbitrary sensing range and distribution, it would be difficult to compute an NE. In order to avoid this computational challenge and make our simulation results evident, we make the following assumptions: (1) All the sensors are identical, and each has a fixed sensing range which is a circle of radius 1.5.


Fig. 5.1. Initial configuration of the network.


Fig. 5.2. Final configuration of the network at iteration 5000 of the DISCL algorithm.
(2) Each point in this region is associated with a uniform value of 1 . With these two assumptions, it is not difficult to see that any configuration where sensing ranges of sensors do not overlap is an NE at which the global potential function is equal to 81.

In this example, the diameter of the location graph is 20 and $N=9$. According to our theoretical result, we should choose an exploration rate of $\epsilon(t)=\left(\frac{1}{t}\right)^{\frac{1}{189}}$. The exploration rate decreases extremely slowly, and the algorithm requires an extremely long time to converge. Instead, we choose $\epsilon(t)=\left(\frac{1}{t+2^{10}}\right)^{\frac{1}{2}}$ in our simulation. Figure 5.1 shows the initial configuration of the group where all of the sensors start at the same position. Figure 5.2 presents the configuration at iteration 5000 , and it is evident that this configuration is an NE. Figure 5.3 is the evolution of the global potential function which eventually oscillates between 78 and the maximal value of 81 . This verifies that the sensors approach the set of NEs.

As in [21], [22], we will use fixed exploration rates in the DISCL algorithm which then reduces to the DHSCL algorithm. Figures 5.4, 5.5, and 5.6 present the evolution of the global potential functions for $\epsilon=0.1,0.01,0.001$, respectively. When $\epsilon=0.1$,


FIG. 5.3. The evolution of the global potential function with a diminishing exploration rate for the DISCL algorithm.


Fig. 5.4. The evolution of the global potential function under $D H S C L$ when $\epsilon=0.1$.


Fig. 5.5. The evolution of the global potential function under $D H S C L$ when $\epsilon=0.01$.


Fig. 5.6. The evolution of the global potential function under DHSCL when $\epsilon=0.001$.


FIG. 5.7. Final configuration of the network at iteration 50000 of the DIACL algorithm.
the convergence to the neighborhood of the value 81 is the fastest, but its variation is largest. When $\epsilon=0.001$, the convergence rate is slowest. The performance of $\epsilon=0.01$ is similar to the diminishing step-size $\epsilon(t)=\left(\frac{1}{t+2^{10}}\right)^{\frac{1}{2}}$. This comparison shows that, for both diminishing and fixed exploration rates, we have to empirically choose the exploration rate to obtain a good performance.
5.2. A numerical example of the DIACL algorithm. We consider a lattice of unit grids, and each point is associated with a uniform weight 0.1 . There are four identical sensors, and each of them has a fixed sensing range which is a circle of radius 1.5. The global optimal value of $U_{g}$ is 3.6 . All the sensors start from the center of the region. We run the DIACL algorithm for 50000 iterations and sample the data every 5 iterations (see Figure 5.7). Figures 5.8 to 5.11 show the evolution of the global function $U_{g}$ for the following four cases, respectively: $\epsilon(t)=\frac{1}{4}\left(\frac{1}{t+1}\right)^{\frac{1}{4}}, \epsilon=0.1$, $\epsilon=0.01$, and $\epsilon=0.001$, where the diminishing exploration rate is chosen empirically. Let us compare Figure 5.9 to 5.11 . The fixed exploration rate $\epsilon=0.1$ renders a fast convergence to the global optimal value 3.6. However, the fast convergence rate comes at the expense of large oscillation. The fixed exploration rate $\epsilon=0.01$ induces the


FIG. 5.8. The evolution of the global potential function under the DIACL algorithm with a diminishing exploration rate.


Fig. 5.9. The evolution of the global potential function under the DIACL algorithm when $\epsilon=0.1$ is kept fixed.


FIG. 5.10. The evolution of the global potential function under the DIACL algorithm when $\epsilon=0.01$ is kept fixed.

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FIG. 5.11. The evolution of the global potential function under the DIACL algorithm when $\epsilon=0.001$ is kept fixed.
slowest convergence but the smallest oscillation. The diminishing exploration rate then balances these two performance metrics. Under the same exploration rate, the DIACL algorithm needs a longer time than the DISCL algorithm to converge. This is caused by the asynchronous nature of the DIACL algorithm.
6. Conclusions. We have formulated a coverage optimization problem as a constrained exact potential game. We have proposed two payoff-based distributed learning algorithms for this coverage game and shown that these algorithms converge in probability to the set of constrained NEs and the set of global optima of certain coverage performance metric, respectively.
7. Appendix. For the sake of a self-contained exposition, we include here some background in Markov chains [16] and the theory of resistance trees [38].
7.1. Background in Markov chains. A discrete-time Markov chain is a dis-crete-time stochastic process on a finite (or countable) state space and satisfies the Markov property (i.e., the future state depends on its present state, but not the past states). A discrete-time Markov chain is said to be time-homogeneous if the probability of going from one state to another is independent of the time when the step is taken. Otherwise, the Markov chain is said to be time-inhomogeneous.

Since time-inhomogeneous Markov chains include time-homogeneous ones as special cases, we will restrict our attention to the former in the remainder of this section. The evolution of a time-inhomogeneous Markov chain $\left\{\mathcal{P}_{t}\right\}$ can be described by the transition matrix $P(t)$, which gives the probability of traversing from one state to another at each time $t$.

Consider a Markov chain $\left\{\mathcal{P}_{t}\right\}$ with time-dependent transition matrix $P(t)$ on a finite state space $X$. Denote by $P(m, n):=\prod_{t=m}^{n-1} P(t), 0 \leq m<n$.

Definition 7.1 (strong ergodicity [16]). The Markov chain $\left\{\mathcal{P}_{t}\right\}$ is strongly ergodic if there exists a stochastic vector $\mu^{*}$ such that for any distribution $\mu$ on $X$ and any $m \in \mathbb{Z}_{+}$, it holds that $\lim _{k \rightarrow+\infty} \mu^{T} P(m, k)=\left(\mu^{*}\right)^{T}$.

Strong ergodicity of $\left\{\mathcal{P}_{t}\right\}$ is equivalent to $\left\{\mathcal{P}_{t}\right\}$ being convergent in distribution and will be employed to characterize the long-run properties of our learning algorithm. The investigation of conditions under which strong ergodicity holds is aided by the introduction of the coefficient of ergodicity and weak ergodicity defined next.

Definition 7.2 (coefficient of ergodicity [16]). For any $n \times n$ stochastic matrix $P$, its coefficient of ergodicity is defined as $\lambda(P):=1-\min _{1 \leq i, j \leq n} \sum_{k=1}^{n} \min \left(P_{i k}, P_{j k}\right)$.

Definition 7.3 (weak ergodicity [16]). The Markov chain $\left\{\mathcal{P}_{t}\right\}$ is weakly ergodic if for all $x, y, z \in X$ and for all $m \in \mathbb{Z}_{+}$, it holds that $\lim _{k \rightarrow+\infty}\left(P_{x z}(m, k)-\right.$ $\left.P_{y z}(m, k)\right)=0$.

Weak ergodicity merely implies that $\left\{\mathcal{P}_{t}\right\}$ asymptotically forgets its initial state, but does not guarantee convergence. For a time-homogeneous Markov chain, there is no distinction between weak ergodicity and strong ergodicity. The following theorem provides the sufficient and necessary condition for $\left\{\mathcal{P}_{t}\right\}$ to be weakly ergodic.

Theorem 7.4 (see [16]). The Markov chain $\left\{\mathcal{P}_{t}\right\}$ is weakly ergodic if and only if there is a strictly increasing sequence of positive numbers $k_{i}, i \in \mathbb{Z}_{+}$, such that $\sum_{i=0}^{+\infty}\left(1-\lambda\left(P\left(k_{i}, k_{i+1}\right)\right)=+\infty\right.$.

We are now ready to present the sufficient conditions for strong ergodicity of the Markov chain $\left\{\mathcal{P}_{t}\right\}$.

Theorem 7.5 (see [16]). A Markov chain $\left\{\mathcal{P}_{t}\right\}$ is strongly ergodic if the following conditions hold:
(B1) The Markov chain $\left\{\mathcal{P}_{t}\right\}$ is weakly ergodic.
(B2) For each $t$, there exists a stochastic vector $\mu^{t}$ on $X$ such that $\mu^{t}$ is the left eigenvector of the transition matrix $P(t)$ with eigenvalue 1 .
(B3) The eigenvectors $\mu^{t}$ in (B2) satisfy $\sum_{t=0}^{+\infty} \sum_{z \in X}\left|\mu_{z}^{t}-\mu_{z}^{t+1}\right|<+\infty$.
Moreover, if $\mu^{*}=\lim _{t \rightarrow+\infty} \mu^{t}$, then $\mu^{*}$ is the vector in Definition 7.1.
7.2. Background in the theory of resistance trees. Let $P^{0}$ be the transition matrix of the time-homogeneous Markov chain $\left\{\mathcal{P}_{t}^{0}\right\}$ on a finite state space $X$. Furthermore, let $P^{\epsilon}$ be the transition matrix of a perturbed Markov chain, say $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$. With probability $1-\epsilon$, the process $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ evolves according to $P^{0}$, while with probability $\epsilon$, the transitions do not follow $P^{0}$.

A family of stochastic processes $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is called a regular perturbation of $\left\{\mathcal{P}_{t}^{0}\right\}$ if the following hold for all $x, y \in X$ :
(A1) For some $\varsigma>0$, the Markov chain $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ is irreducible and aperiodic for all $\epsilon \in(0, \varsigma]$.
(A2) $\lim _{\epsilon \rightarrow 0^{+}} P_{x y}^{\epsilon}=P_{x y}^{0}$.
(A3) If $P_{x y}^{\epsilon}>0$ for some $\epsilon$, then there exists a real number $\chi(x \rightarrow y) \geq 0$ such that $\lim _{\epsilon \rightarrow 0^{+}} P_{x y}^{\epsilon} / \epsilon^{\chi(x \rightarrow y)} \in(0,+\infty)$.

In (A3), $\chi(x \rightarrow y)$ is called the resistance of the transition from $x$ to $y$.
Let $H_{1}, H_{2}, \ldots, H_{J}$ be the recurrent communication classes of the Markov chain $\left\{\mathcal{P}_{t}^{0}\right\}$. Note that within each class $H_{\ell}$, there is a path of zero resistance from every state to every other. Given any two distinct recurrence classes $H_{\ell}$ and $H_{k}$, consider all paths which start from $H_{\ell}$ and end at $H_{k}$. Denote by $\chi_{\ell k}$ the least resistance among all such paths. Now define a complete directed graph $\mathcal{G}$ where there is one vertex $\ell$ for each recurrent class $H_{\ell}$, and the resistance on the edge $(\ell, k)$ is $\chi_{\ell k}$. An $\ell$-tree on $\mathcal{G}$ is a spanning tree such that from every vertex $k \neq \ell$, there is a unique path from $k$ to $\ell$. Denote by $G(\ell)$ the set of all $\ell$-trees on $\mathcal{G}$. The resistance of an $\ell$-tree is the sum of the resistances of its edges. The stochastic potential of the recurrent class $H_{\ell}$ is the least resistance among all $\ell$-trees in $G(\ell)$.

ThEOREM 7.6 (see [38]). Let $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$ be a regular perturbation of $\left\{\mathcal{P}_{t}^{0}\right\}$, and for each $\epsilon>0$, let $\mu(\epsilon)$ be the unique stationary distribution of $\left\{\mathcal{P}_{t}^{\epsilon}\right\}$. Then $\lim _{\epsilon \rightarrow 0^{+}} \mu(\epsilon)$ exists, and the limiting distribution $\mu(0)$ is a stationary distribution of $\left\{\mathcal{P}_{t}^{0}\right\}$. The stochastically stable states (i.e., the support of $\mu(0)$ ) are precisely those states contained in the recurrence classes with minimum stochastic potential.

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[^1]:    ${ }^{1}$ See [5] for a definition of proximity graph.

